



US Army Corps of Engineers Little Rock District AD-A224 869

RED RIVER WATERSHED

GILLHAM LAKE

COSSATOT RIVER, ARKANSAS

EMBANKMENT CRITERIA AND PERFORMANCE REPORT



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OCTOBER 1987

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EMBANKMENT CRITERIA
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PERFORMANCE REPORT

October 1987

U. S. Army Engineer District, Little Rock Corps of Engineers

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Prepared by
Grimes & Johnson, Inc.
Little Rock, Arkansas
and
Grubbs, Garner & Hoskyn, Inc.

Little Rock, Arkansas



GILLHAM DAM AND RESERVOIR COSSATOT RIVER, ARKANSAS

EMBANKMENT CRITERIA AND PERFORMANCE REPORT

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GILLHAM DAM AND RESERVOIR SALINE RIVER, ARKANSAS

EMBANKMENT CRITERIA AND PERFORMANCE REPORT

PERTINENT DATA

LOCATION

Cossatot River, Mile 49, Howard and Polk County, Arkansas.

PROJECT STRUCTURES

Rockfill dam with compacted earth core, concrete gravity spillway controlled by tainter gates, and intake structure with facilities for a flood control conduit, low flow outlet, water supply outlet, an office building, and recreational facilities.

PROJECT PURPOSES

Flood control, water supply, water quality control, and fish and wildlife conservation.

DRAINAGE AREA

Two hundred seventy one (271) square miles upstream from the dam.

ELEVATIONS, AREAS AND STORAGES

	:	Elevation	:	Areas	:	St	ora	ige
Feature	:(1	feet, NGVD)	:	(Acres) <u>:</u> .	Acre-feet)	:(Inches)(1)
Top of Dam	:	586.0	:	6,000	:	-	:	-
Maximum Pool	:	581.0	:	5,590	:	283.310	:	19.60
Top Flood Control Pool	:	569.0	:	4,680	:	221,780	:	15.34
Spillway Crest	:	527.0	:	2,330	:	78,580	:	5.44
Top Conservation Pool	:	502.0	:	1,370	:	33,030	:	2.28
Top Inactive Pool	:	464.5	:	310	:	3,720	:	0.26
Flood Control Storage	:	502.0-569.0	:	-	:	188,750	:	13.06
Conservation Storage	:	464.5-502.0	:	-	:	29,310(2):	2.03
•						, ,	•	

⁽¹⁾ From total drainage area of 271 square miles.

⁽²⁾ Includes 20,600 acre-feet for water supply and 8,700 acre-feet for water quality control.

EMBANKMENTS

Location Cossatot River at mile 49.0

Type Nonoverflow embankment

Type of Fill Random rock with impervious core

Slope Protection Riprap upstream and downstream

Height 160 feet above streambed

Length 1,750 feet

Crest Width 32 feet

Top Elevation 586.0 feet NGVD

Design Flood Maximum probable flood

Freeboard Five feet above maximum pool

Used for Roadway Yes

Tlevation of Streambed 426.0 feet NGVD

SPILLWAY

Location Right abutment about 1,500 feet west of

main embankment

Type Gated concrete gravity ogee weir

Crest Elevation 527.0 NGVD

Net overflow Width 200 feet

Number and Size of Gates Four - 50' x 41'

Type of Gates Tainter

Top of Gate Elevation 569.0 NGVD

Induced Surcharge None

Design Head 54.0 feet

Maximum Discharge Capacity 272,700 c.f.s.

Bridge Deck Elevation 586.0 NGVD

Type Energy Dissipator Flip bucket

Time Required to Open/

Close all Gates 42 minutes

OUTLET FACILITIES

Intake Structure One intake structure with facilities for flood

control conduit, water supply, and low flow

intakes

FLOOD CONTROL CONDUIT

Location Near west end main embankment - Right Abutment

Purpose Flood Control

Type of Outlet Concrete-lined tunnel

Size of Outlet 10' diameter

Type of Service Gate Hydraulically-operated slide gates

Number and size of Gates Two 4.5' x 10'

Entrance Invert Elevation 437.0 NGVD

Minimum Time Required to

Open/Close Service Gate 10 minutes

Type Emergency Closure

and Time Required Hydraulically-operated slide gate requiring

10 minutes to close gate.

Type Energy Dissipator Concrete stilling basin with baffles

LOW FI.OW OUTLET

30" Pipe

Location Near west end main embankment

Purpose Low flow releases

Type of Outlet Circular pipe

Size of Outlet 30-" diameter at vent pipe

Type of Service Gate Hydraulically-operated butterfly valve.

Original insert gates located in main

service gates were deleted in 1986.

Multilevel Intake

Elevation 487.0 and 472.0 NGVD

WATER SUPPLY OUTLET

Location Near west end main embankment

Purpose Water Supply

Type of Outlet Circular pipe

Size of Outlet 24" diameter

Type of Service Gate Gear-operated gate valve

Number and Size of Gate One - 24 inches

Multilevel Intake Elevations 477.0 and 458.5 NGVD

1. INTRODUCTION

1.1 <u>Purpose and Scope of Report</u>. This report provides a summary record of significant design, construction and operational data on the Gillham Dam and Reservoir. It was prepared in accordance with ER 1110-2-1901, "Embankment Criteria and Performance Report," dated 31 December 1981 and is for use by engineers to familiarize themselves with the project, re-evaluate the embankment when needed, and for guidance in design of comparable future projects. This report was prepared jointly by Grimes & Johnson, Inc. and Grubbs, Garner & Hoskyn, Inc., for the Soils and Materials Section, Little Rock District, under the provisions of Contract No. DACW38-86-D-0070, Delivery Order No. 0007.

The report presents a general description of the foundation conditions, type of material and placement methods of the various sections of the embankment, the design considerations on stability and seepage control, significant operational events, and an evaluation of the conditions of the embankment. Pertinent drawings, design and construction data, and photos are also included. More detailed descriptions of the foundations conditions are contained in the report "Gillham Dam Foundation, Part III, Embankment and Foundation Grouting."

1.2 <u>Brief Description and Purpose of Project</u>. The Gillham Dam and Reservoir are located in Howard and Polk Counties, Arkansas. The dam is at mile 49.0 on the Cossatot River, a tributary of the Little River in southwest Arkansas. It is approximately 6 miles northeast of the town of Gillham, Arkansas and 75 miles north of Texarkana, Arkansas.

The dam consists of an embankment, a gated spillway and outlet works. The main dam embankment is a 1,750 foot long rockfill dam with compacted earth core rising to 160 feet above the streambed. Located on the right abutment in a natural saddle, is a gated concrete gravity ogee weir spillway 200 feet wide, with 4 - 50-foot tainter gates. The outlet works has a 10-foot diameter conduit (concrete lined tunnel) controlled by hydraulically operated slide gates in the gate tower with releases made through the stilling basin at the downstream portal.

There is also a 30" diameter low flow outlet and a 24" diameter water supply outlet. The outlet works is located at the right abutment of the main dam. The service bridge to the gate tower parallels the main dam from the right abutment overlook area. Spur dikes, saddle dikes, and a training dike are located as shown on Plate 2. Other structures include an office/maintenance building and recreational facilities.

The project purposes are flood control, water supply, water quality, and fish and wildlife conservation. Recreation was not a designated project purpose; however, it was later added as incidental to them.

- 1.3 <u>Project Authorization</u>. The Gillham Dam and Reservoir project was authorized by the Flood Control Act approved 3 July 1958 (Public Law 85-500, 85th Congress, 2nd Session). Related legislation included House Document No. 170, the report "Miliwood Reservoir and Alternate Reservoirs, Little River, Oklahoma and Arkansas."
- 1.4 Project Construction History. Construction of the right abutment access road, Howard County road relocations, and the project buildings began in June 1963 and was completed in September 1966. Construction of the outlet works began in January 1966 and was completed in June 1968. Construction commenced in April 1968 on the spillway and was halted in February 1971. Fork resumed in 1972 and was completed in August of that year. The contract for construction of the main embankment was awarded in July 1972 and completed in October 1975. Photographs taken during construction are contained in Appendix B. Impoundment of the lake began in May 1975 and reached conservation pool elevation of 502.0 NGVD in April 1976. The pool of record occurred on 5 December, 1982 at elevation 561.64 NGVD.

The public use areas at Gillham Lake were constructed between June 1975 and July 1976. They include Cossatot Reefs, Cossatot Point, Coon Creek, and Little Coon Creek.

A summary of construction contracts for the entire project is contained on the following page.

LIST OF PROJECT CONTRACTS

CONTRACT NO.	CONSTRUCTION FEATURE	CONTRAC		COMPLETION DATE
DA-34-066 CIVENG-65-1726	Right Abutment Access Road	\$ 394,22	3 20 Jun 63	22 Sept 66
Contractor:	G. F. Thomison Const., P.	0. Box 15	, Eufaula, OK	
DA-34-066 CIVENG-65-1728	Construction of Project Buildings	\$ 134,7	30 11 Mar 65	02 Jun 66
Contractor:	LaForge & Budd Co., P. O.	Box 393,	Parsons, KS	
DA-34-066 CIVENG-66-223	Construction of Outlet Works	\$1,901.0	000 14 Jan 66	07 Jun 68
<u>Contractor:</u>	Martin K. Eby Const., 610	N. Main S	t., Wichita, l	KS 67201
DACW56-66-C-0384	Relocation of Howard Co Roads	\$ 558,0	55 11 May 66	17 Nov 67
Contractor:	E. W. Blair Inc., P. O. B	ox 450, Br	oken Bow, OK	
DACW56-68-C-0183	Construction of Dikes & Spillway	\$2,925,8	59 26 Apr 68	01 Jul 72
Contractor:	Martin K. Eby Const., 610	N. Main S	st., Wichita,	KS 67201
DACW56-73-C-0008	Construction of Main Embankment	\$4,131,7	725 31 Jul 72	10 Oct 75
Contractor:	Hixson & Lehenbauer, P. 0	. Box 1574	, Topeka, KS	66601
DACW56-73-C-0217	Addition to Intake Structure	\$ 129,4	122 11 Jun 73	17 Jul 74
Contractor:	Amis Const. Co., P. O. Bo	x 1871, Ok	City, OK	
DACW56-73-C-0242	Overlook Structure No.1	\$ 98,3	300 14 Jun 73	10 Apr 74
Contractor:	LaForge & Budd Co., P. O.	Box 393,	Parsons, KS	
DACW56-75-C-0209	Construction of Public Use Areas Stage I (Cossaatot Point, Cossatot reefs, Coo			
Contractor:	Triark Inc. & Souter & As	soc., P. O). Box 1052, L	R, AR 72203
DACW56-76-C-0104	Repair of Structural Cracks & Weir Surface of Spillway	\$ 29,68	30 N/A	N/A
Contractor:	Hunt Process Corp.	<u>,</u>		

1.5 <u>Periodic Inspections</u>. In accordance with the requirements for periodic inspection and continuing evaluation of completed civil works structures, the following have been published for periodic inspections that have been performed on the embankment, spillway and outlet works at Gillham Dam and Reservoir. They contain the problems developed after construction and instrumentation data.

Periodic Inspection Report No. 1 April 1975
Periodic Inspection Report No. 2 April 1976
Periodic Inspection Report No. 3 November 1979
Periodic Inspection Report No. 4 April 1983
Periodic Inspection Report No. 5 March 1985

2. GEOLOGY

- 2.1 General Geology and Topography. The area of the project is a part of the rugged Ouachita Mountain section of the Ouachita physiographic province. Bedrock consists of intensely folded Paleozoic rocks. As a result of the erosion of these folded strata, the general area is characterized by parallel hogback ridges separated by narrow, steep-sided valleys. Trellis drainage patterns are common. Streams are swift and flood plains are poorly developed or absent.
- 2.2 Geology of Site. At the Gillham site, the Cossatot River, flowing southward, has cut through an east-west trending ridge to form a arrow, steepwalled valley. The configuration of the dam site and that of the spillway site, located in a saddle about 1,700 feet northwest of the right abutment, is shown on Plate 2. Bedrock is the Stanley shale of Mississippian-Pennsylvanian age. This formation consists of hard quartzitic sandstones and hard silty to sandy shales. The overburden mantle, which usually masks the bedrock ever on steep slopes, has an average thickness of approximately 6 feet along the dam axis and about 4 feet in the spillway area. The greatest thickness of overburden was in Hole No. 31 near the base of the right abutment where talus more than 31 feet thick was found. Talus forms a large part of the overburden on the steep slopes, especially on the right abutment. The overburden is mostly residual silt and sand. There is almost no flood plain, and very little alluvium at the site.

The water table was near river level in the right abutment (Hole No. 37), whereas in the left abutment the water table was observed to rise to about 20 feet above the level of the river (Hole No. 34). Several seeps and small springs were observed about 10 feet above the river at the base of the left abutment. In the spillway area, the water table was found at elevation 515, about 25 feet below the surface, at the low point of the saddle (Hole No. 28).

The principal structural feature in the site area is an eastward plunging syncline whose axis is near the axis of the dam and roughly parallel to it. The spillway site is on the north flank of the syncline and, since the syncline is plunging, the rocks strike at an angle of about 30° to the

spillway axis. Therefore, both downstream and right-to-left dip components are formed. The thick shale which underlies the left side of the spillway site has been eroded from higher elevation to form the notch or saddle in which the spillway is located. The shale was found to exist at each end of the outlet tunnel (Plate 13). Local contortions which affect both dip and strike occur often and are particularly noticeable in this shale.

2.3 Foundation Description.

d.

- 2.3.1 Spillway. Bedrock for the spillway consists of both shale and sandstone. A line drawn diagonally from the downstream right corner through the upstream left corner of the apron would be close to the contact of the sandstone and shale at design grade. Upstream and to the right of this line, the foundations for the weir, right nonoverflow and half of the apron will be predominantly sandstone. Downstream and to the left of the diagonal line most of the left nonoverflow and the other half of the apron is underlain The sandstone is hard, fine-grained quartzitic and contains many Joints are abundant and are shaly zones, shale beds and shale partings. usually open and water stained. Partial or complete drill-water loss was experienced in most of the holes drilled in the sandstone. A few of the joints contain calcite or quartz fillings. The silty to sandy shale which forms part of the spillway foundation has been metamorphosed to a hard brittle slickensided rock. All bedding planes are smooth or slickensided and sometimes have a polished appearance. Where unweathered, the shale is not readily susceptible to air slaking. The rocks in the spillway area strike in a direction of N 67° E, or parallel to the diagonal line across the apron, mentioned above, and dip to the southeast (downstream and from right to left) at an angle of 22° (40 feet per hundred). The top of firm rock, as shown on the log sections on Plate 6, was found to vary from 13 to 33 feet in depth.
- 2.3.2 Outlet Works. The shale and sandstone in the outlet works area are similar to the shale and sandstone described for the spillway. The shale which was excavated in the approach and outlet channels of the outlet works is the same member which forms the foundation for the left nonoverflow section of the spillway. The sandstone overlying the shale, which is apparently continuous almost from portal to portal of the tunnel, is more massive with fewer and less prominent shale beds than the stratigraphically

lower spillway sandstone. Lenticular beds of both shale and sandstone occur near the contact of the two members, as illustrated by the logs of Holes Nos. 38 and 39. As shown on Plate 13, dips were into the tunnel from both ends. Based on nearest outcrops, dip components parallel to the axis of the tunnel amount to 19° at the downstream end and 11 at the upstream end.

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- 2.3.3 Embankment. The flood plain is very narrow, the valley is steep-sided and the main stream flows on a rock bottom. The overburden along the axis is quite thin, attaining a maximum thickness of only 8.8 feet as determined in Core Hole 33 located near the base of the right abutment. overburden consists primarily of low plasticity silts (ML) with some basal deposits of gravelly clays. A geological profile along the axis of dam showing the sparse overburden cover is shown on Plate 3. Bedrock along the cutoff trench for the dam is sandstone containing minor amounts of shale lenses. Most of the shale is found in the left abutment. Both shale and sandstone, like the rock in general at the Gillham site, are slightly metamorphosed and are both hard competent rocks. Some of the sandstone is intensely jointed with open joints sometimes only inches apart. portion of Hole No. 37, located on the right abutment, penetrated several feet of such rock. Firm rock varies from the top of rock under the river to depths of nearly 20 feet below the top of rock in the abutments. The presence of many closely spaced open joints in the upper rock usually accounts for the deep top of firm rock in the abutments.
- 2.4 <u>Investigations</u>. Twenty core holes were drilled and nine test pits were excavated by backhoe at the embankment. Twenty-three core holes were drilled along the outlet works abutment and forty-nine core holes were drilled in the spillway area. Results of foundation borings are contained on Plates 3 through 5.

3. FOUNDATION TREATMENT

á

3.1 Design Considerations.

- 3.1.1 Embankment. Bedrock along the cutoff trench for the dam consists predominantly of sandstone and lesser amounts of shale. The rock is hard and slightly metamorphosed but the sandstone contains many open and tight water stained joints. Top of the firm rock was near top of rock under the river, as shown on Plate 3. Consequently, a cutoff trench to firm rock or to a depth of no greater than 10 feet below top of rock was selected from about station 23+20A on the right abutment to about station 38+80A on the left abutment. The trench design had a 25 foot bottom width below elevation 550 and a 15 foot bottom width between elevations 550 and 569 at the ends of the cutoff trench. Where the foundation rock contained many open joints resulting in depths to firm rock in excess of 10 feet below top of rock, a layer of grout or concrete was used to cover the bottom of the trench to protect the core from piping. The same treatment was to be used for open joints in the bottom of the trench where the trench is excavated to firm rock. Because of the extensive jointing which occurs above the firm rock line, the downstream slope of the trench was covered with gunite. A grout curtain was designed in the bot, m of the cutoff trench from end to end of the dam below the flood control pool level. The grout curtain for the embankment would overlap and center between two rings of curtain grout for the outlet works tunnel.
- (a) <u>Underseepage Probabilities</u>. Underseepage was not considered to be a major problem due to the limited thickness of soil cover at the site; however, to insure a strong foundation which is essential for a dam embankment over 150 feet in height, the foundation under the entire embankment was stripped to rock. Excavation of rock was confined to a 25 foot wide cutoff trench along the centerline. The cutoff trench was backfilled with impervious clay soils.
- (b) <u>Settlement Conditions</u>. Appreciable magnitudes of settlement were not anticipated, particularly in view of the absence of any thick compressible clay horizons. Most of the anticipated minor settlement would occur during the construction period, at least to the extent that overbuilding of the embankment was not considered warranted.

3.1.2 <u>Spillway</u>. Vertical or near vertical faces against which concrete sections were placed, were either line drilled or presplit. The shale faces were protected with either a bituminous sealing compound or pneumatically applied concrete. The sandstone required no protective coating. Rock bolts were installed to prevent undercutting of the rock face at the right end of the weir section with wire mesh to prevent rock from falling to the lower level. Rock slopes in sandstone were as steep as 4:V on 1:H or steeper except on the approach channel, where slopes of 1:V on 1:H were selected to conform to the dip. Slopes in shale were also cut to 1:V on 1:H. The design included rock anchors installed to a depth of 15 feet in rock beneath the weir section and to a depth of 10 feet beneath the stilling basin slab.

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A grout curtain was provided from the gallery beneath the spillway and a series of drain holes were drilled from and vented into the gallery, (Plate 31). Drain holes extended 10 ft. into rock beneath the spillway slab. A cutoff trench about 10 feet into rock and 15 feet wide extended from each end of the nonoverflow monoliths to the ends of the embankment. A grout curtain was provided over the length of the cutoff trench.

3.1.3 <u>Outlet Works</u>. In the outlet works area, the sandstones were cut to steep slopes, 4:V on 1:H or steeper, with shales cut to 1:V on 1:H slopes. Vertical or near vertical faces, against which concrete was placed, were formed by line drilling or presplitting. Shale faces exposed longer than a few days were covered by a bituminous sealing compound on the steep slopes and with protective concrete on the horizontal or near horizontal faces. Rock bolts and wire mesh were used both in the tunnel and on the steep slopes of the open cuts. Steel ribs were installed at the tunnel portals and wherever necessary for tunnel support. Seepage, although expected in the tunnel, was not considered to create any major problem.

To insure a solid contact between the concrete liner and the rock, contact grouting in the crown of the tunnel was required, and consolidation grouting was performed to fill the fissures and voids in the rock, (Plate 48). Two rings of curtain grouting were placed 10 fe.t apart near the center of the tunnel at the axis of the dam. The ring curtain grouting was to overlap the embankment grout curtain centered between the grout rings (Plate 48).

3.2 Embankment Design.

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3.2.1 Main Embankment.

- (a) <u>General</u>. A revised typical section for the embankment is shown on Plates 14 and 16. The zoning revisions were made primarily to provide for greater use of quarry-run rock as well as more efficient use of the other available borrow materials.
- (b) <u>Impervious Material</u>. The core zone was not changed from initial design. It was constructed of clays and plastic silts (CL, CH, and ML) having a liquid limit exceeding 20 and a percent of fines of 60 or more. The fill was allowed to contain scattered stones with a maximum size of 3 inches.
- (c) <u>Random Fill</u>. Zones of random materials were added both upstream and downstream of the core zone through the revisions. The width of the downstream zone was increased approximately 11 feet to use more of the available random material. These zones, which are essentially the same as the impervious zones of the initial design, consist of materials having a percent of fines of 40 or more. Maximum size stones within the random fill material did not exceed 6 inches.
- (d) <u>Filter A Material</u>. A design change was made during construction which consisted of the 3 foot zone of fine filter (filter A) being moved downstream 11 feet as a result of increasing the width of the random fill. Filter A consisted of concrete sand.
- (e) <u>Transition Material</u>. Design changes during construction in the dcwnstream transition zone consisted of eliminating the horizontal portion and including a small zone of filter C material to transition the filter A zone to the rockfill A zone. The gradation of filter C material is as follows:

	Percent by
Sieve size	Weight passing
6 inch	100
4 inch	90-100
2 inch	65-85
1 inch	40-70
3/8 inch	15-35
No. 4	0-10

The upstream transition material remained essentially the same except that the material was placed between the random fill and random rockfill zones. Widths of the transition zones varied between 8 and 12 feet, as shown on Plates 14 and 16, to use as much as possible of the by-product obtained from the production of rockfill A.

- (f) <u>Rockfill</u>. The upstream and downstream rockfill shells were rezoned, as shown on Plate 14, to use more rockfill B (random) and less rockfill A (processed). This was due primarily to a better quality of stone for the rockfill B zones. Specifications for these materials are as follows:
- (1) <u>Rockfill B</u>. Fresh or weathered quarry-run sandstone with a maximum size of 200 pounds and not more than 15 percent by weight of friable (highly weathered) sandstone and shale. The material was placed in 18 inch lifts and compacted with four passes of a 10 ton vibratory or 50 ton pneumatic roller.
- (2) <u>Rockfill A</u>. Fresh, processed sandstone with a maximum size of 250 pounds, except for the outer 5 feet of the upstream shell which may contain 350 pound maximum size stone. All rockfill A materials were processed to remove shale and other materials smaller than 3 inches. After processing the finished product did not contain more than 5 percent by weight of shale or material smaller than 3 inches.
- (g) <u>Stability</u>. Revised stability analyses for the embankment design are presented on Plates 21 through 26. Minimum safety factors obtained for these analyses of the revised embankment are shown in Table 1.

TABLE 1

EMBANKMENT SAFETY FACTORS

	:	Shear	:		:			
	:	strength	:	Type of	:_	Safet	y fa	ctors
Condition	:	used	:	analysis	:	Computed	_:	Required
	:		:		:		:	
End of construction	:		:		:		:	
Upstream	:	Q	;	Arc	:	1.55	:	1.3
	:	Q	:	Wedge	:	2.17	:	1.3
Downstream	:	Q	:	Arc	:	1.77	:	1.3
	:	Q	:	Wedge	:	2.32	:	1.3
Sudden drawdown	:	R	:	Arc	:	1.17	:	1.0
Steady seepage	:	R	:	Arc	:	1.49	:	1.5
	:	S	;	Arc	:	1.50	:	1.5
	:	R+S/2	:	Wedge	:	2.05	:	1.5
	:		:		:		:	

The results of the stability analyses indicate the revised embankment section is satisfactory.

3.2.2 Cofferdams.

- (a) General. The cofferdams for construction of the outlet works were inconsequential (see Design Memorandum No. 10, Outlet Works). However, the closure cofferdam (Cofferdam No. 2) (Plate 14) constituted a significant portion of the embankment. To use the cofferdam to the maximum possible extent as a part of the permanent embankment, the position and configuration presented in General Design Memorandum No. 4 was revised by Supplement No. 1 to Design Memorandum No. 9. The cofferdam (see typical section on Plate 14) was over 100 feet in height, and except for temporary layers on the upstream slope, consists of portions of all zones of the embankment below El. 525. The upstream cofferdam slope was set equal to that of the permanent embankment (1:V on 2:H). The downstream slope was also 1:V on 2:H although not extending to the ultimate width of the downstream rockfill section. In deference to the potential for overtopping and related loss of embankment materials, a cable anchorage system was provided to protect the rockfill sections. downstream cofferdam (No. 3) was minor in height (8 feet maximum) and posed no unique problem.
- (b) <u>Wire Mesh System</u>. It was anticipated that the central cofferdam No. 2 would be overtopped during construction. If overtopping and

subsequent breaching occurred, the Government would not only be faced with the cost and time delays in making repairs, but with significant increases in quantity and cost of fill materials. The price increases would result as follows:

- (1) The upstream borrow areas contain only minor quantities of reserve fill materials, not enough to replace a major washout.
- (2) An alternate borrow source about 7 miles downstream on private lands would involve truck haul over county and access roads, making the cost of these materials about 5 times more than those obtained from the upstream borrow areas.
- (3) The second alternative of replacing a portion of the soil materials with rockfill would increase the cost by a factor of 4.

To engage large rock masses in resistance to the erosion or drag forces that would develop during overtopping a wire mesh reinforcement system was selected (Plate 9). The mesh would function similar to a Gabion system where small rock is tied together to act as a single unit. The specifications required that the crest profile be maintained level to prevent concentration of flow and that the contractor be prepared to cover the downstream lip with reinforcement whenever overtopping was considered imminent.

The District investigated the possibility of leaving a portion of the cofferdam low to serve as a spillway but found that the increased depth of flow and the interference with construction made it more desirable to allow overflow over the full length. The control elevation was located at the lip of the downstream slope. The reinforced cofferdam slope was contained within the finished embankment, and therefore it was not considered necessary to remove or overlay the reinforcement.

With the above alternatives in mind, the addition of an anchored wire mesh system was felt warranted in the revision of 1970 (Plate 18). This system offered greater resistance to the potential overflow forces than individual stones. Subsequently, in 1973 the final modification to the slope protection

for Cofferdam No. 2 was adopted after loss of embankment material during overtopping (Plate 17).

- 3.2.3 <u>Impervious Materials</u>. Borrow was limited to those areas located in the flood plain upstream from the dam axis, all within the fee purchase limits. Estimates based on explorations indicated sufficient materials were available to meet the basic requirements for the impervious portions of the embankment. The soils for the select impervious zone were in short supply. As shown in Table 1, Appendix A, these soils were confined to borrow area C, although alternate sources were available at increased costs.
- 3.2.4 <u>Rockfill</u>. Seven individual ledge outcrops of quartzitic sandstone were explored in the search for an adequate and suitable quarry site for the source of rockfill material. An estimated 5,500,000 cubic yards of usable rockfill material could be excavated from the portions of the ledge where the overburden was not excessive. A typical sample of the quartzitic sandstone in the quarry was tested and the results of the physical tests are listed in Table 2, Appendix A.
- 3.2.5 <u>Transition and Filter Materials</u>. The transition material was produced from the waste at the rockfill quarry. The sand filter layer was obtained from off-site sources.

3.2.6 Spillway.

- (a) <u>Spillway Wraparounds</u>. A section of the spillway wraparound is shown on Plate 2. The section was designed as a rockfill shell with an impervious earth core. The impervious zone was placed against the concrete with a minimum horizontal dimension of 5 feet near the crest and a 1:V on 1:H slope to the base. A layer of coarse material was placed between the impervious core and the rock shells to form a transition. The maximum exterior slope was 1:V on 2.5:H. All soil was removed from the wraparound foundation areas. Spur dikes, placed on either side of the spillway entrance, were constructed of spillway rock excavation with exterior slopes of 1:V on 2.5:H.
- (b) Embankment. The embankment design was the same as the main dam embankment as presented.

(c) <u>Saddle Dikes</u>. A typical section of the saddle dikes is shown on Plate 29. The dikes attained a maximum height of only 12 feet and were an all earth section rather than rockfill as used in the main dam. This was a more economical design because of the near proximity of earth materials in comparison to the distant location of rock either from the required excavation or the quarry site. The slopes were flattened to 1 on 6 to eliminate the need for riprap and backing. An inspection trench was excavated to rock along the centerline of Dike No. 2.

3.2.7 Outlet Works.

- (a) <u>Tunnel</u>. The 10-foot-diameter concrete lined tunnel began 48 feet downstream from the downstream face of the gate tower. The tunnel was designed in accordance with EM 1110-2-2901, using case IV. Supporting steel ribs were used during excavation where needed for 1 of support and remain as reinforcing for the tunnel lining. Sections of the tunnel are shown on Plates 46, 47 and 48.
- (b) <u>Stilling Basin</u>. The stilling basin extends downstream from the outlet portal and provides for a reduction in velocity of the tunnel discharge. Immediately downstream from the outlet portal, the structure consists of a U-3haped channel with vertical walls which flare out to the full width of the stilling basin slab. The walls and slab were anchored to the adjoining rock by grouting anchor bars into drilled holes. Drains were provided behind the walls and beneath the slab to reduce the hydrostatic pressure. The unbalanced hydrostatic uplift for the stilling basin slab was reduced by 50 percent as recommended by EM 1110-2-2400. Details of the stilling basin are shown on Plate 49.
- 3.2.8 <u>Compaction Control</u>. Compaction control criteria given in the specifications for the embankment zones are presented in Tables 3.2.8(a) and 3.2.8(b). The Test Embankment Report prepared by Tulsa District is contained in Appendix C.

TABLE 3.2.8(a)

COMPACTION CONTROL CRITERIA FOR EMBANKMENT MATERIALS

	:_			Туре	of M	ateri	al			
Item	:	Select	:				:_	Roc	ckf	111
	: I	mperviou	s:	Impervious:	Fil	ters	: W	eathered	:	Fresh
	:		:	:			:		:	
Roller, types	:		:	:			:		:	
and number of	:		:	:			:		:	
passes	:		:	:			:		:	
Tamping	:	8	:	8 :	Prohi	bited	1:P	rohibited	d:P	rohibited
Pneumatic,	:		:	:			:		:	
50-ton,	:P	rohibite	d:	6 :	3		:	4	:P	rohibited
Vibratory,	:		:	:			:		:	
10-ton,	:P	rohibite	d:	Prohib!ted:	(1)3		:P	rohibited	d:	4
Tractor,	:		:	:			:		:	
crawler	: P	rohibite	d:	Prohibited:	3		:P	rohibite	d:P	rohibited
	:		:	:			:		:	
Lift thickness,	:		:	:			:		:	
loose inches	:	8	:	(2)8 :	12		:	12	:	24
	:		:	:			:		:	
Target densities	:	95	:	95 :	70		:	(4)	:	(4)
(3)	:		:	:			:	•	:	- •

- (1) A lightweight steel wheel roller was allowed for the sand filter.
- (2) Twelve inches, if pneumatic roller was used.
- (3) In terms of standard AASHO method except for filter criteria which refers to relative density.
- (4) Methods of control were identical to those used for test fill. Periodic check tests (by one cubic yard test pits) were also used to determine density and gradations of in-place rockfill.

TABLE 3.2.8(B)

MOISTURE CONTROL CRITERIA FOR EMBANKMENT MATERIALS

	:	Range in Moisture,
Zone	<u> </u>	Percent from wo
	:	
elect Impervious	:	0 to +4
	:	
mpervious	:	-3 to +3
	:	
ilter	:	None (i)
	:	

⁽¹⁾ Filter materials were moistened in accordance with the latest revisions to the guide specification. Sluicing was prohibited.

4. FOUNDATION CONSTRUCTION PROCEDURES

4.1 Embankment Foundation.

- 4.1.1 Overburden Excavation. Hixson and Lehenbaur, Inc., of Topeka, Kansas, the prime contractor, began clearing the embankment area on 19 September 1972. Stripping of overburden material began on 19 October 1972. Suitable material was placed in cofferdam No. 1. Unsuitable material was used to construct haul roads and a river crossing or wasted in designated areas. This excavation was accomplished with the use of three D-8 Caterpillar dozers, two 621 Caterpillar scrapers, one 988 Caterpillar loader, and two 769 Caterpillar tail dumps.
- 4.1.2 Embankment. A cutoff trench was presplit and excavated to a depth of 10 feet from station 23+15 to Station 27+80 and station 31+50 because top of rock was firm. Total planned width of the cutoff trench was 25 feet. However, the trench was widened and deepened in the loose fractured rock encountered between station 26+00 and station 27+50. Where the foundation rock had many open joints in the bottom of the trench, the joints were filled with concrete or grout to protect the core from piping. The downstream slope of the trench from station 25+90 to station 27+30 and station 32+20 to station 39+75 was covered with a 2" coat of gunite because of extensive jointing. A grout curtain was formed along the dam axis between station 23+20 and station 39+40. A single line of holes, varying in depth from 20 feet to 100 feet, was drilled and grouted. The holes were angled 25 into the left and right abutments with a transition between station 32+40 and station 33+30. The grout curtain for the embankment overlaps and centers between two rings of curtain grout provided for the outlet works tunnel.
- 4.1.3 <u>Treatment</u>. Surfaces to receive materials other than impervious or random fill did not require special foundation preparation. Blade cleaning of these surfaces as part of the common excavation was adequate. Treatment of the foundation rock within the limits of the random fill and impervious zones consisted of cleaning all rock surfaces, open joints, and fractures by washing, air jet, or both. Open joints in the rock surface were sealed and capped with grout or mortar. Small fractures and open joints in the rock were cleaned to

the depths directed and covered with cement mortar or thick grout. Large fractures and large open joints in the bottom of the cutoff trench were cleaned out to a depth directed by the Contracting Officer and filled with concrete. Exposed final rock surfaces which were subject to deterioration were immediately covered and protected by a layer of the overlying embankment material.

- 4.1.4 <u>Slide Area</u>. During the night of 16 January 1973 a slide of approximately 15,000 cubic yards of insitu material occurred in a portion of the right abutment. The slide encompassed an area some 200 feet square from station 23+90 to station 25+90 immediately downstream of the cutoff trench. This block of material moved down dip into the excavated cutoff trench with an apparent pivot point approximately 200 feet downstream of centerline at approximately station 25+80. (See photographs 1 through 5). A thin clay shale seam, immediately above the floor of the cutoff trench and lubricated by percolating groundwater, provided the slippage plane. Removal of the slide material began on 6 February 1974 and was completed 13 February 1974. Approximately 12,000 cubic yards were removed and incorporated in the embankment rockfill section.
- 4.1.5 Overexcavation. Considerable widening of the cutoff trench was required on the lower portion of the right abutment. Because of the extremely fractured, loose condition of the rock encountered from station 25+00 to station 27+50 the cutoff trench was widened as well as deepened. See paragraph 4.2.2. After consultation with District Office personnel, it was decided to remove the upstream and downstream faces until suitable rock was reached. In any case, the widening was not carried beyond the limits of the toe of the random fill. This widening varied from 20 feet to 70 feet upstream and 40 feet to 60 feet downstream of the planned centerline. Total plan width of the cutoff trench was 25 feet. Some minor overexcavation also occurred on the left abutment along the upstream face where the dip slope was intercepted by the cutoff trench causing some sloughing along the dip slope.

4.2 Grout Curtain.

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4.2.1 <u>Drilling and Grouting</u>. The grout curtain was formed by drilling and grouting a single line of holes, varying in depth from 20 to 100 feet,

along the dam axis from station 23+20 to station 39+40. Generally the holes were angled 25 into the left and right abutments with the transition occurring from station 32+40 to station 33+30. The primary holes (20-foot centers) were drilled to full depth, washed, pressure tested, and grouted in accordance with the "stop grouting, split spacing" method. The secondary holes were located midway between the primary holes. This split spacing continued until the holes were determined to be tight. The drilling and grouting was done by the Judy Company, a subcontractor. The Judy Company began operations on 15 January 1973 and completed their work on 28 April 1974; however, work was not continuous within these dates (See paragraph 4.2.2). Equipment used is listed in Table 4.2.1. Photograph 18 shows grouting operations.

TABLE 4.2.1

GROUTING EQUIPMENT

<u>Item</u>	Quantity
Chicago - Pneumatic 600 Air Compressor	1
Gardner - Denver 900 Air Compressor	1
Chicago - Pneumatic AirTrac Drill	1
Gardner - Denver AirTrac Drill	1
Gardner - Denver Grout Pump	1
Wilden Air Pump, 2 inch	1
Deming Air Pump, 3 inch	1
Truck, 2-ton	1
Pickup, 1/2-ton	2
Parts Trailer	1
Assorted Valves, Gages, Hoses	

4.2.2 <u>Difficulties Encountered</u>. The slide mentioned previously occurred during the night of 16 January 1973 forcing Judy Company to delay operations approximately one month. (See photograph 2.) Operations resumed 19 February with drilling grout holes at station 30+40. The valley bottom and left abutment was grouted with no d'fficulty. On 28 May 1973 grouting operation began on the right abutment and problems were experienced. It was originally planned to form the grout curtain from station 26+50 to station 27+00 by drilling a fanshaped pattern of holes from station 26+20 to station 26+40. Since pressure testing resulted in considerable water losses above 15 feet, grout was introduced by gravity flow. After approximately 30 minutes grout began to leak from a spring 40 feet upstream at approximately station 27+80. After pumping 874 sacks of cement with no apparent tightening of the foundation, the grouting operation was suspended temporarily on 12 June 1973.

The cutoff trench was deepened approximately 8 feet by drilling and blasting. This exposed a large fracture partially filled with grout at station 26+45 to 26+65. (See photographs 28, 29 and 30.) The decision was made to fill this area with concrete. Since routine grouting methods were not successful in this area, a Field Order was issued directing the Contractor to drill and sand grout a line of holes on 1-foot centers 16 feet deep, 10 feet upstream of centerline from station 26+23 to 26+68, to form a plug between the grout curtain along centerline and the open joints and fractures upstream. Work on the Field Order began 9 July 1973 and was completed 18 July 1973. On 24 July 1973 the grout curtain was completed to the base of the slide area.

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Judy Company resumed operations on 18 February 1974. Another problem area was encountered from station 25+20 to station 25+70. See photographs 19, 20, 23, 24, 25, 39, 40 and 42.) Two large open fractures extended across the cutoff trench at station 25+30 and station 25+50. The larger of these, a V-shaped opening 7 - 10 feet deep and 2 - 3 feet wide at the top at station 25+30 also extended upstream under the abutment in excess of 65 feet. The accessible portions were filled with high slump concrete and an attempt was made to grout the centerline. Since grout was bypassing the concrete plug, 13 holes 18 feet deep were drilled upstream of the cutoff trench to intersect the fracture. A thick 3:1 sand grout was mixed in ready-mix trucks and a total of 26 cubic yards were poured into these holes (see photographs 21 and 22), with an additional 3 cubic yards poured into the grout holes on centerline at station 25+30 (1/2 cubic yard) and station 25+50 (2-1/2 cubic yards). centerline had been grouted to refusal, additional rows of holes 10 feet and 5 feet upstream were drilled and grouted to insure that the area was tight (See photograph 41). As a final check, two holes were drilled 15 and 35 feet upstream of the face to intersect the fracture. These holes were tight. Grouting was completed 28 April 1974.

- 4.2.3 Grouting Summary. Table 4.2.3 summarizes the embankment foundation grouting totals for the job. A breakdown of data included in Table 4.2.3 shown in Table 4.2.4 by stations and zones.
- 4.2.4 <u>Conclusions</u>. The sandstone and shale comprising the foundation of the embankment were firm and fresh and considered adequate to support the imposed load. The entire length of the grout curtain for the full depth was

pressure grouted to refusal. Subsequent high water retained by the cofferdam produced no unusual increase in seepage downstream. These facts indicate that the grout curtain is performing the design function.

TABLE 4.2.3

GROUTING SUMMARY

Drilling (ft)	Pres. Test	Sacks of Cement	<u>Pumptime</u>	Connections
19,085	71 hr. 57 min.	7,347	386 hr. 33 min.	523

TABLE 4.2.4

GROUT TAKE SUMMARY

	Zone 1	Zone 2	Zone 3			acks Cement
Stations	(below 40')	(15'-40')	(2'-15')	Total	Drilled	Per Foot
		Primary	r Wolo			
		-	enters)			
		(20 00	incors,			
23+20-27+90	74.2	17.9	280.7	372.8	1905	0.20
28+00-31+95	301.2	142.8	29.8	473.8	2000	0.24
32+00-39+40	10.5	541.6	494.5	1046.6	2950	0.35
		Secondar	w Holes			
		(10' Ce				
		(10 00	,			
23+20-27+90	47.6	25.6	890.1	963.3	2100	0.46
28+00-31+95	122.3	30.2	28.4	180.9	2000	0.09
32+00-39+40	9.4	93.4	227.0	329.8	2900	0.11
		Tertiary	r Holes			
		(5' Cen				
		(0 00	,			
23+20-27+90	3.7	66.8	806.5	877.0	880	1.00
28+00-31+95	0.0	6.6	27.9	34.5	848	0.04
32+00-39+40	-	325.1	70.4	395.5	600	0.66
		Quaternar	v Holes			
		(2.5' Ce				
23+20-27+90	_	47.5	676.4	723.9	665	1.09
28+00-31+95	-	-	-	-	-	-
32+00-39+40	-	0.0	86.0	86.0	208	0.41
		Quinterar	w Woles			
			centers)			
23+20-27+90	, mag	7.2	168.8	176.0	320	0.55
28+00-31+95		-	-		-	-
32+00-39+40	-	0.0	0.0	0.0	80	0.00

Holes (0.63' Centers)

23+20-27+90	-	-	0.0	0.0	80	0.00
28+00-31+95	-	-	-	-	-	-
32+00-39+40	-	-	-	-	-	-
		Grout Tak	e By Zones			
23+20-27+90	125.5	165.0	2822.5	3113.0		
28+00-31+95	423.5	179.6	86.1	689.2		
32+00-39+40	19.9	960.1	877.9	1857.9		
	568.9	1304.7	3786.5	5660.1		
	Grout pum	ped before de	epening			
C.O.T Station 26+20 to 26+40				874.0	sacks	
	Grout and Station	365.4	sacks			
	Backfill	and allowable	waste	447.5	sacks	
		TOTAL		7347.0	sacks	

- 4.3 <u>Spillway</u>. Vertical or near vertical faces against which concrete was placed were either line drilled or presplit. All shale faces were covered with double spray applications of bituminous protection. Brecciated shale was removed along the faulted sandstone-shale contact. The foundation beneath the spillway structure and the cut-off trench for the wrap-around embankments were grouted. Grout holes in the cut-off trench were drilled vertically on 10 feet centers. Grout holes in the spillway gallery were inclined upstream at 28 measured from vertical as shown on Plate 12. Grout holes were formed through the concrete by black steel pipe installed on 5 feet centers and extending from the foundation to the gallery gutter. The grouting profile is also shown on Plate 12. Foundation drains were installed on the downstream side of the gallery to depths of 35 feet and 50 feet. Drain holes were spaced 10 feet on centers and were inclined downstream at an angle of 20° measured from vertical.
- 4.4 <u>Outlet Works</u>. Excavation faces were formed by presplitting and close line drilling. Shale faces were covered with a bituminous sealing compound on the steep slopes and with protective concrete on horizontal or near horizontal faces. Rock bolts and wire fabric were placed throughout the tunnel except in the transition section and areas containing steel rib supports. Steel ribs

were also installed from station 5+60T to station 6+23T and station 10+78T to station 10+96T on 3 feet centers for a total of 29 supports. A concrete tunnel liner was installed and contact grouting was performed to provide a solid contact between the concrete liner and the rock. After the contact grouting was completed the voids in the rock surrounding the bore were grouted. Two rings of curtain grout were placed 10 feet apart near the center of the tunnel at the axis of the dam. The ring curtain grouting overlaps the embankment grout curtain which was centered between the grout rings.

5. CONSTRUCTION CONTROL DATA.

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- 5.1 <u>Construction Sequence</u>. The general construction sequence for the main dam embankment was as follows:
 - (a) Construct cofferdams No. 1 and No. 3.
 - (b) Construct log boom.
 - (c) Grout and excavate foundation.
 - (d) Construct cofferdam No. 2.
 - (e) Complete embankment and associated work.

5.2 Embankment Fill.

- 5.2.1 <u>Typical Section</u>. A typical section of the main embankment is shown on Plate 16. The maximum height of the main embankment is 162 feet. The section is essentially symmetrical with a central impervious core, random fill zones flanked by transition material, and outer shell zones of rockfill. A three foot thick inclined filter A is connected to the downstream rockfill zone by a transition zone and coarse filter C material.
- 5.2.2 Borrow. Earth materials for the impervious and random zones were obtained from borrow areas A1, A2, B and C. Additional borrow materials were obtained from borrow area E for construction of embankment and dikes in the spillway area. Additional borrow for construction of the main embankment was obtained from Hunter's Creek downstream of the dam site and from Striklin Creek located upstream along the left abutment. The soils ranged from fine-grained to granular types. The predominant soil was a very lean sandy clay, low in plasticity, essentially borderline between a clayey silt and a clayey sand (CL, sandy but also includes ML-CL, sandy and some SC.) Gravel and shale were present in the lower elevations of most areas. A summary of borrow soil test results is contained in Appendix A.
- 5.2.3 <u>Filter</u>. Materials for the inclined filter (filter A), and filter C were obtained from the southern portion of borrow area B. These materials, primarily gravels, were processed to the extent necessary to meet the specifications.

5.2.4 Rockfill and Transition Materials. The quarry site is located upstream from the dam site and about 5,000 feet east of the river. Prior to construction of the embankments, a test fill was built to develop engineering data for materials (Rockfill A and Rockfill B) from the quarry site. Details of the test embankment are shown in Appendix C, in the Test Embankment Report, Gillham Dam, prepared by Tulsa District. Material for the Rockfill "B" zones consisted of Class I or II rock with a maximum size of 200 pounds with less than 15 percent by weight either passing the No. 4 screen, or consisting of friable (highly weathered) stone, or shale material, in any load. Material for Rockfill "A" consisted of Class I rock with a maximum size of 250 pounds with the exception that stone placed in the outer five feet of the upstream Rockfill "A" zone had a maximum size of 350 pounds and an intermediate size that 40% of the stone weighed more than 50 pounds. Processing was required to remove shale and materials smaller than three-inch size and this by-product was used for the transition material.

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During construction, three in-place density determinations were made for the Rockfill "A" and Rockfill "B" materials. The results of the tests are shown below.

ROCKFILL A

Test No.	Station	<u>Offset</u>	<u>Elevation</u>	<pre>Dry Density #/ft3</pre>	Minus 3"
1	31+07	147'D.S.	439.1	113.2	28
2	29+85	138'D.S.	463.1	120.8	25
3	31+00	80'U.S.	540.0	116.3	16.9

ROCKFILL B

Test No.	<u>Station</u>	<u>Offset</u>	Elevation	Dry Density	Minus 3"
				#/ft3	%
1	31+08	220'U.S.	443.9	138.6	15.8
2	28+00	130'U.S.	474.0	136.6	13.0
3	33+00	140'D.S.	500.0	136.4	11.0

The results of the Rockfill density tests indicated that the in-place densities were some 5% to 15% greater than the design unit weights for Rockfill B and A respectively.

5.3 Embankment Stability.

5.3.1 Tests and Design Values. Shear tests were conducted on representative soils in the remolded state for impervious and random soils. Test results from test fills at Gillham Dam and Dierks Dam were used in determining the unit weights. Results of tests of borrow materials are contained in Appendix D, results of tests of borrow soil are contained in Appendix E, and results of tests of rock, rockfill, riprap and concrete aggregate are contained in Appendix F. The Adopted Design Strengths are shown in the following tabulation:

			Adopted Design Strengths					
	Saturated	Submerged	(Q .	1	3	S	;
	Unit Weight	Unit Weight	ø	C	ø	С	ø	C
Materials	(pcf)	(pcf)	(deg)	(tsf)	(deg)	(tsf)	(deg)	(tsf)
Rockfill A (1)	135	72.5	42	0.	42	0.	42	0.
Rockfill B (2)	135	72.5	36	0.	36	0.	3ó	0.
Transition (3)	125	62.5	33	0.	33	0.	33	0.
Filter (3)	125	62.5	33	0.	33	0.	33	0.
Randomfill	125	62.5	5	0.5	22	0.2	28	0.
Impervious	125	62.5	5	0.5	22	0.2	28	0

Ø=Angle of internal friction C=Cohesion

- (1) Material assumed to have 40 percent voids and a saturated surface dry unit weight of 105 pcf. Angle of internal friction was assumed to be 42°.
- (2) Angle of internal friction was assumed to be 36°.
- (3) Angle of internal friction was assumed to be 33°.
- 5.3.2 <u>Construction Control Tests</u>. The following summary contains the results of the construction Record Samples from the impervious and random fill zones as compared to the design strengths of these materials. A graphical presentation of the shear test data is presented on Plate 20.

CONSTRUCTION CONTROL SHEAR TESTS

		Test St	rength	Design Strength		
No.	Type	Ø	C	ø	С	
Tests	Test	Degrees	t.s.f.	Degrees	<u>t.s.f.</u>	
8	Q	10	0.9	5	0.5	
8	R	24	0.6	22	0.2	
8	s	34	0.0	24	0.0	

The results of the construction control in place density, water contend, and classification tests for the random and impervious materials are summarized below:

CONSTRUCTION CONTROL FIELD TESTS

Material			Compaction %	Wf-Wo %	Percent Fines	LL %
		Average	99.3	-0.10	66.4	27.3
Impervious	(1)	High	103.8	+2.2	95.1	44.3
-		Low	95.2	-2.5	52.0	18.0
		Average	99.4	+ .04	60.2	25.1
Random	(2)	High	106.6	+2.9	92.2	35.0
		Low	94.5	-2.6	39.5	17.0

- (1) Results of 82 tests.
- (2) Results of 163 tests.

The results of the field tests indicate the field moisture content of the random and impervious materials was essentially within the limits specified with only a small number of the tests having a moisture content slightly drier than specified.

5.3.3 Stability. The embankment was analyzed for stability under conditions of end of construction, sudden drawdown, partial pool and steady seepage. The trial slip surfaces were analyzed using the procedures set forth in Engineering Manual, EM 1110-2-1902, 1 April 1970. The following tabulations indicate the safety factors obtained for the most critical slip surface based on the design shear strengths and the "as built" shear strength of the random and impervious materials. The lowest safety factors obtained for each condition analyzed are shown in the following tabulation:

	Shear Strength	Type of	Safe	ty Fac	tors
Condition	Used	Analysis	As Built	Design	Required
End of Construction (US)	Q	Arc	1.66	1.55	1.3
End of Construction (US)	Q	Wedge	1.66	2.17	1.3
End of Construction (DS)	Q	Arc	1.86	1.77	1.3
End of Construction (DS)	Q	Wedge	1.86	2.32	1.3
Sudden Drawdown (2)	R	Arc		1.17	1.0
Sudden Drawdown (1)	S,R	Arc	1.28		1.0
Partial Pool	S,(R+S)/2	Arc	1.52		1.5
Steady Scopage (4)	R	Àrc		1.49	1.5
Steady Seepage (4)	S	Arc		1.50	1.5
Steady Seepage (3)	S,(R+S)/2	Arc	1.70		1.5
Steady Seepage (4)	(R+S)/2	Wedge		2.05	1.5
Steady Seepage (3)	S,(R+S)/2	Wedge	1.56		1.5

(See Notes, Next Page)

- (1) Analysis for sudden drawdown from maximum pool elevation 581.0 to conservation pool elevation 502.0. The rockfill "A" zone was assumed to be free draining.
- (2) Analysis for sudden drawdown from flood control pool elevation 569.0 to conservation pool 502.0. The rockfill "A" zone was assumed to be free draining.
- (3) Analysis based on differential head between maximum water surface elevation 581.0 and maximum tailwater elevation 480.0.
- (4) Analysis based on differential head between flood control pool elevation 569.0 and maximum tailwater elevation 480.0.
- 5.3.4 <u>Conclusion</u>. Based on a comparison of test results of the in-place materials with those used in design, the stability of the structure is considered adequate.

5.4 Design Modification.

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- 5.4.1 Right Abutment Slide. A slide containing approximately 15,000 cubic yards of insitu material occurred on the right abutment between sta. 23+90 and sta. 26+25 on 16 January, 1973. The slide resulted from excavation of the cutoff trench and moved along a bedding plane at the base of the cutoff trench. Removal of the slide material eliminated the downstream face of the cutoff trench through this reach and deletion of the 2-inch thick coating of pneumatically placed concrete. Approximately 12000 cubic yards of the slide mass was incorporated into the embankment sections.
- 5.4.2 Revised Design-Cofferdam No. 2. Cofferdam No. 2 was overtopped five times during the construction period. The first three of these overtoppings resulted in substantial losses of fill materials. The initial cofferdam design is shown on Plate 18 and the revised design on Plate 17. The original design had the timber anchorages on the same horizontal plane as the wire mesh it was holding and therefore when the overlying material was washed away there was no ballast to protect the timber anchorages. Basically, the revised design consisted of two inverted skillet weir sections which provided an anchorage system to hold the wire mesh and cable system in place when the slopes were subjected to overtoppings. An additional feature was the placement of an airfield landing mat, and a wire wrapped rockfill zone at the

toe of the weirs to provide a stilling effect to prevent erosion at the toe of the slope and to absorb the shock of the abraded material.

5.4.3 Earth Fill. As identified as a possible consequence during design, the cofferdam overtoppings did in fact result in a loss of earth fill materials of approximately 200,000 cubic yards. As a result, earth borrow materials became a critical supply item and particularly those materials meeting the impervious specification requirements of liquid limit exceeding 20 and percent fines exceeding 60. To relieve this shortage, the random and impervious zones were combined into an earth fill material with a minimum liquid limit of 22 and percent fines of at least 35. The moisture and compaction controls were not changed. This modification resulted in the use of the sandy silty clays in the central core that were previously excluded because of the fines restriction. The material loss also required the use of borrow material from borrow areas A-2, C, Stricklin Creek and Hunters Creek that had not been anticipated.

6. INSTRUMENTATION.

6.1 <u>Surface Control Monuments</u>. To monitor the embankment movements a total of 25 embankment surface control monuments and 8 abutment control monuments have been established. The location of the monuments are shown on Plate 19.

During construction, monitoring procedures consisted generally of surveying all the completed monuments at the time a new line of monuments were installed, hence readings were taken as follows:

<u>Date</u>	Embankment C.L. Elev.	Remarks
17 July 1974	540	Initial Line 4 readings
28 August 1974	560	Initial Line 3 readings
10 September 1974	561	Initial Line 1 reading
27 September 1974	564	
10 December 1974	586	Initial Line 2 readings

6.2 Embankment Movements. The data through 27 September 1974 revealed minimal movements; however, measurements made on 10 December 1974 indicated that horizontal movements of 6 to over 12 inches upstream had occurred between During the period of 27 September - 10 December 1974, the embankment had been raised about 22 feet in height, requiring approximately cubic yards of embankment materials. During this same period a total of 15.21 inches of rain fell on the watershed and the reservoir rose above elevation 475+/- four times with the maximum rise to about elevation 490 The Resident Office advised the District occurring on 25 November 1974. Office of the 10 December 1974 readings and an immediate program of obtaining weekly horizontal and monthly vertical readings was initiated. monitoring data showed that subsequent measured changes in monument location were within the error of observation of the survey methods (generally 0.1 to 0.2 feet), significant additional fill movements had apparently not occurred by March 1975. Four temporary monuments were also located upstream of surface monument line 1 on 18 December 1974. These temporary monuments have shown virtually no movement. Some bulging of the upstream slope is apparent in the vicinity of the line of temporary monuments; however, field personnel reported that this is due to a construction slope correction and had been evident prior to July 1974. Temporary iron pins were driven on 18 December 1974 along the dam axis to monitor vertical movements of the maximum dam section.

maximum settlement of 0.15 feet (station 28+04 and 30+01), in the same area where the large horizontal movements occurred, was measured 5 February 1975. Detailed information is contained in the report "Gillham Dam, Report of Embankment Slope Movements," by Tulsa District, dated March 1975.

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- 6.3 <u>Hypothesis of Movements</u>. Several hypotheses as to the reasons for the movements have been drawn, among these are: point crushing and quarry dust removal due to wetting of the embankment and lateral spreading.
- 6.4 <u>Conclusion</u>. It appears that the construction phase movements at Gillham Dam were not caused by a design or construction deficiency, but are a result of several phenomena that occur naturally in all earth and earth-rock dams to some degree. Lateral spreading and slope material rearrangement were both very likely contributors to the movements that occurred at the Gillham Dam. The lack of sufficient monitoring data to determine the chronology of the movements was significant in that there is no definitive way of directly correlating or excluding physical events such as rainfall and pool rises as potential catalysts.
- 6.5 <u>Post-Construction Embankment Movements</u>. Periodic inspections through No. 5 in March 1985 reported the embankment to be in excellent condition with no significant changes between inspections in 1979, 1983 and 1985. Maximum measured embankment settlement from initial survey December 1974 through August 1984 has been approximately 1.2 feet (near Sta. 27+00 and Sta. 33+00). This value is less than the overbuild of 2 feet.

Minor seepage only has occurred, and was attributed to discharge from the outlet works (Periodic Inspection Report No. 3).

APPENDIX A

TABLES

TABLE 1 SUMMARY OF BORROW SOILS - CHARACTERISTICS AND ESTIMATED QUANTITIES

-

			Estim	Estimated Amounts of Soil (cu. yd.)	of Soil :	Soj	Soil Characteristics	arac	teri	stic	l i i
Borrow : Area :	Approximate Location	Estimated : Total : Quantities:	Percent: F :(Percent: Percent F : Percen F : (Impervious):(Select	Percent F : Select	••	3 "		rercent r	ent	<u>.</u>
••		: (cu.yd.):	(0-40):	(0-40); (40-60) :I.	:Impervious :: (60+) ::	:Max:Min:Avg:Max:Min:Avg : : : : :	fin:A	.vg:N	lax:M :	1n:}	Sv.
A North	Right bank, 1 mile upstream	217,000:	5,000:	60,000:	152,000:	33:	15:	24:	75:	42:	59
A South	: Right bank, 1 mile upstream	218,000:	30,000:	129,000:	59,000:	27:	14:	22:	:99	38:	51
.	Left bank, 1 mile upstream	420,000:	83,000:	114,000:	223,000:	28:	17:	24:	. 89	36:	22
C West	Right bank, 2 miles upstream	70,000:	30,000:	21,000:	19,000:	31:	8	22:	78:	33:	53
C East	Right bank, 2 miles upstream	120,000:	:	5,000:	115,000:	40:	20:	29:	87:	:09	73
D(abandoned):			• •• •	• •• ••	••••	•••	•••	•• ••	•• ••	•• ••	
E West (1)	Right abutment, 1.5 miles downstream	350,000:	24,000:	116,000:	210,000:	33:	16:			50:	63
E East	Right abutment, 1.5 miles downstream	100,000:	47,000:		53,000:	33:	20:	28:	: 98	37:	65
ĨZ4	Right bank, 4 miles downstream	500,000:	20,000:	185,000:	295,000:	29:	19:	24:	78:	53:	29
_U	Right bank, 6 miles downstream	2,550,000:	535,000:	918,000:	1,097,000:	27:	14:	19:	74:	50:	26
•	Total	4,545,000; 774,000	774,000:	1,548,000:	2,223,000:					•• •• ••	
		•	•	,				-			-

Portions of horrow area E have been made available for constructing the right abutment access road now under contract. (1)

TABLE 2

PHYSICAL TESTS ON QUARRY SANDSTONE

Test	:	Test result	
	:		
Bulk specific gravity, SSD	:	2.63	
	:		
Unit weight, p.c.f.	:	164.1	
	:		
Absorption, percent	:	0.8	
	:		
Soundness, percent loss	:		
Mg SO4, 5 cycles	:	1.1	
Freeze-thaw, 25 cycles	:	1.1	
	:		
Abrasion, L.A., percent loss	:		
A grading	:	22.1	
B grading	:	15.4	

APPENDIX B

PHOTOGRAPHS

GILLHAM DAM AND RESERVOIR SALINE RIVER, ARKANSAS

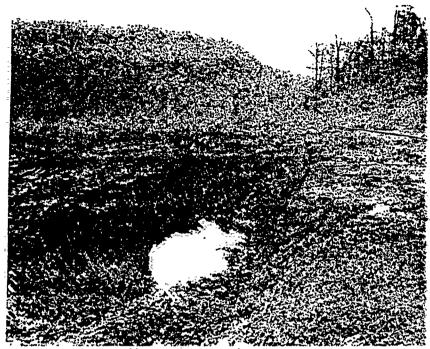
EMBANKMENT CRITERIA AND PERFORMANCE REPORT

APPENDIX B - PHOTOGRAPHS

Page No.	Photo No. and Description
B-1	Photo No. 1. Slide area from Sta. 23+90 to 24+90, D/S of centerline, as viewed from Sta. 23+00.
B-1	Photo No. 2. View of the upper end of the slide looking D/S from Sta. 24+00. Judy Company (grouting subcontractor) is removing water and air lines.
B-2	Photo No. 3. A view of the slide looking down station from Sta. 24+00.
B-2	Photo No. 4. The slide area as viewed form Sta. 27+00 looking uphill.
B-3	Photo No. 5. The silde area after removal of loose material as viewed from STa. 23+50.
B-3	Photo No. 6. Overall view of the foundation across the valley floor as seen from the outlet tower.
B-4	Photo No. 7. A view of the cut-off trench (C.O.T.) above Sta. 25+20. Note grout pipes along centerline.
B-4	Photo No. 8. An overall view of the right abutment along centerline. The fill crosses centerline at Sta. $26+50(+/-)$.
B-5	Photo No. 9. Overall view of the right abutment. The strip of clean foundation occurs at Sta. 27+50 to 27+70.
B-5	Photo No. 10. View SW along a minor fault which crosses centerline at approx. Sta. 28+40.
B-6	Photo No. 11. Foundation under random and impervious zones form Sta. 30+50 to 31+80 as seen from upstream toe of the random zone at Sta. 31+80.
B-6	Photo No. 12. View of the foundation from STa. 28+60 to 30+50 along centerline as seen from Sta. 27+00.
B-7	Photo No. 13. Foundation under impervious zone from Sta. 33+30 to 33+75 on left abutment.
B-7	Photo No. 14. Foundation under impervious and U/S random zones at Sta. 34+40. Note shale-sandstone contact.
B-8	Photo No. 15. Foundation under impervious and U/S random zones at Sta. $35+60$ to $36+00$. Note extent to which the U/S face has been removed.
B-8	Photo No. 16. Foundation under D/S random zone from Sta. 35+50 to 35+70.

- B-9 Photo No. 17. Foundation under the impervious zone from Sta. 34+40 to 38+65. Note the evidence of presplitting on the U/S (left Hand) face.
- B-9 Photo No. 18. Grout pump used on this contract. Mixing tank is being charged. The mixed grout is emptied into the holding tank to be pumped into the grout holes.
- B-10 Photo No. 19. Open fracture at Sta. 25+30 as seen looking D/S from centerline.
- B-10 Photo No. 20. Open fracture at Sta. 25+30 as seen from centerline loock U/S.
- B-11 Photo No. 21. Placing ready-mixed sand grout in holes drilled U/S of C.O.T. to intersect open fracture at Sta. 25+30.
- B-11 Photo No. 22. View of grout holes U/S of C.O.T. at Sta. 25+30 as seen looking U/S along strike of the fracture.
- B-12 Photo No. 23. D/S portion of the open fracture at Sta. 25+30 prior to backfill with concrete.
- B-12 Photo No. 24. Extreme D/S end of the fracture at the D/S toe of the random zone.
- B-12 Photo No. 25. Open fracture at STa. 25+30 being backfilled with concrete.
- B-13 Photo No. 26. D/S half of the step face at Sta. 25+80(+/-) prior to placing concrete to eliminate the overhang.
- B-13 Photo No. 27. U/S half of the step face at Sta. 25+80(+/-) priot to placing concrete to eliminate the overhang.
- B-14 Photo No. 28. Large open fracture at Sta. 26+40 to 26+60 partially filled with grout. Openings below and to the left and right of the hard hat extend into the U/S face for an unknown distance.
- B-14 Photo No. 29. Large open fracture at Sta. 26+40 to 26+60 partially filled with grout. An opening to the right of the hard hat extends into the D/S face for an unknown distance. In the lower right-hand corner of the photograph a coke can plugs core Hole #37 at elevation 484. Note grout pipes in foreground to be installed in openings prior to backfilling with concrete.
- B-15 Photo No. 30. Placing foundation protection concrete at Sta. 26+40 to 26+60 the temporary form was used to aid in running concrete into the opening in the face.
- B-15 Photo No. 31. A partially grout-filled fracture at Sta. 26+40. This is the D/S extension of the large fracture at Sta. 26+40 to 26+60. Clean up prior to backfilling with concrete.
- B-16 Photo Nc. 32. This photograph shows the upstream extension of the fracture at Sta. 26+40 to 26+60. Note the concrete (placed 7-2-73) plug in the background.
- B-16 Photo No. 33. This view (looking U/S from centerline) of the fracture at Sta. 26+40 to 26+60 shows partial filling with grout at the lower right. This area was backfilled with concrete.

- B-17 Photo No. 34. Clean-up prior to application of "Gunite" to D/S face of C.O.T. in left abutment. Note gunite machine mounted on the flatbed trailer.
- B-17 Photo No. 35. "Gunite" being applied to D/S face of C.O.T. in left abutment.
- B-18 Photo No. 36. "Gunited" D/S face and step face at Sta. 26+90 to 27+10(+/-).
- B-18 Photo No. 37. A small protable mixer was used to produce concrete to eliminate small overhangs. This photo was taken at Sta. 26+40(+/-) D/S of centerline.
- B-19 Photo No. 38. Sand grout is being mixed by hand to backfill a small crack extending D/S from the previously concreted area at 26+50(+/-). The concrete plug lies under the broom at the right edge of the photo.
- B-19 Photo No. 39. A view within the right abutment of the fracture 35' U/S of centerline.
- B-19 Photo No. 40. A view within the right abutment of the fracture 55' U/S of centerline.
- B-20 Photo No. 41. Foundation along centerline at Sta. 25+50 to 25+70. Note three lines of grout holes in this area.
- B-20 Photo No. 42. Foundation from Sta. 25+50 to 25+70 as seen from a point 40 feet D/S. The open fracture at left is to be backfilled with concrete.



Slide area from Sta 23+90 to 24+90. D S of centerline, as viewed from Sta 23+00.



View of the upper end of the slide looking I/S from Sta 24+00. Judy Company (grouting subcontractor) is removing water and air lines.



Photo. No. 3

E view of the slide looking down station from Sta 24+00.



Photo. No. 4

The slide area as viewed from Sta 27+00 looking uphill.



Photo. No. 5

The slide area after removal of loose material as viewed from Sta 23+50.



Photo. No. 6

Overall view of the foundation across the valley floor as seen from the outlet tower.

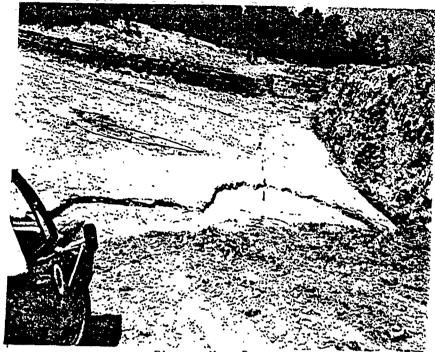


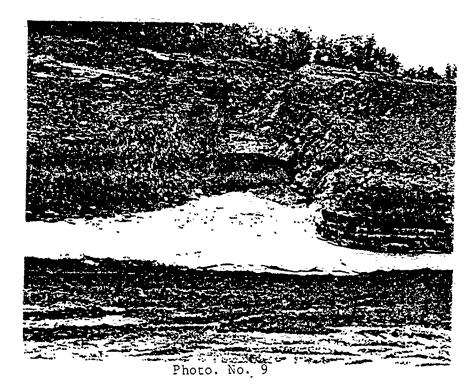
Photo. No. 7

 \pm view of the C.O.T. above Sta 25 \pm 20. Note grout pipes along Centerline.

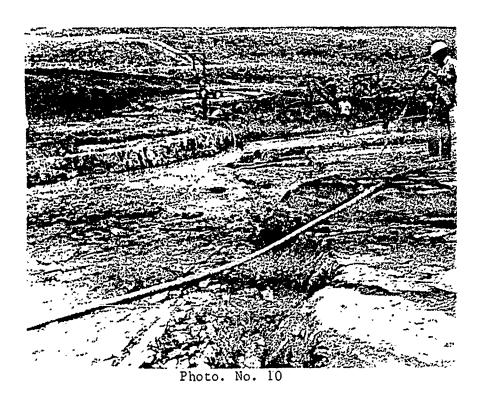


Photo. No. 8

An overall view of the right abutment along centerline. The fill crosses centerline at Sta $26 \div 50 \pm .$



Overall view of the right abutment. The strip of clean foundation occurs at Sta $27{\pm}50$ to $27{\pm}70$



View SW along a minor fault which crosses centerline at approx. Sta $28\!+\!40$.

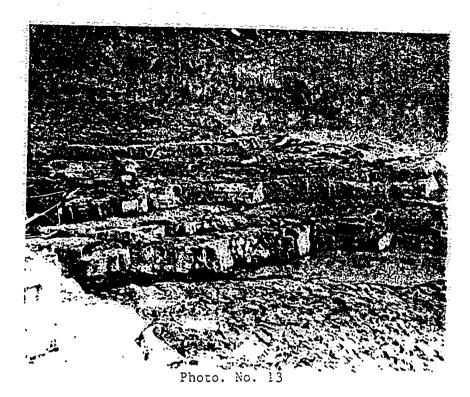


Founation under random and impervious zones from Sta 30 ± 50 to 31 ± 80 as seen from upstream toe of the random zone at Sta 31 ± 80 .



Photo. No. 12

View of the foundation from Sta 28+60 to 30+50 along centerline as seen from Sta 13+00.

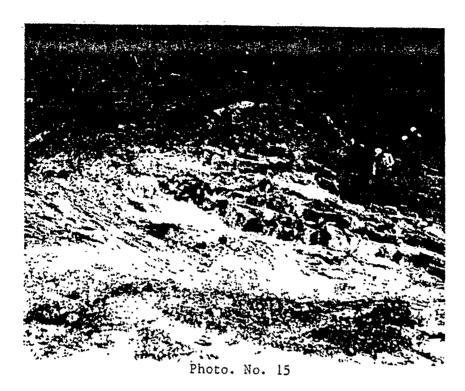


Foundation under impervious zone from Sta 33÷30 to 33÷75 on left abutment.



Photo. No. 14

Foundation under impervious and U/S random zones at Sta 34+40. Note shale-sandstone contact.



Foundation under impervious and U/S random zones at Sta 35+60 to 36+00. Note extent to which the U/S face has been removed.



Foundation under D/S random zone from Sta 35+50 to 35+70.



Photo. No. 17

Foundation under the impervious zone from Sta 1:-- 1: 2:- 5. Note the evidence of presplitting on the U.S. (left want last

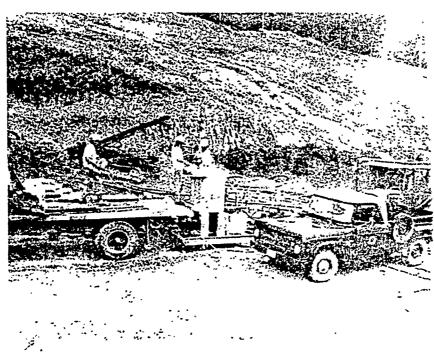


Photo. No. 18

Grout pump used on this contract. Mixing tank is being energed. The mixed grout is emptied into the helding term to the grout holes.



.... t A &: Fra 15-30 as seen looking D/S from centerline.



Photo. No. 20

The structure at 122 25-30 as seen from centerline looking U/S.

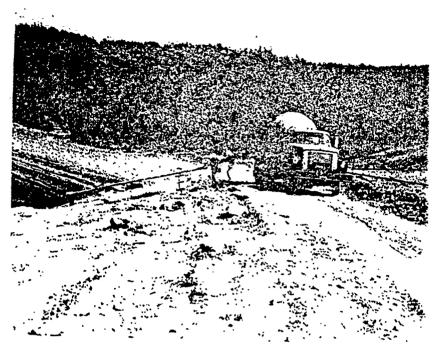


Photo. No. 21

Placing ready-mixed sand grout it halve unlaw U S of C.C.T. to intersect open fracture at Sus 19-3 .



Photo. No. 22

View of grout holes U/S of C.c.T. at Sta 25+30 as seen looking U/S along strike of the fracture.

D/S portion of the open fracture at Sta 25+30 prior to backfill with concrete.



Photo. No. 23

Extreme D/S end of the fracture at the D/S toe of the random zone.



Photo. No. 24

Open fracture at Sta 25÷30 being backfilled with concrete.



Photo. No. 25



Photo. No. 26

D/S half of the step face at Sta $25 \div 80 \div$ prior to placing concrete to eliminate the overhang.

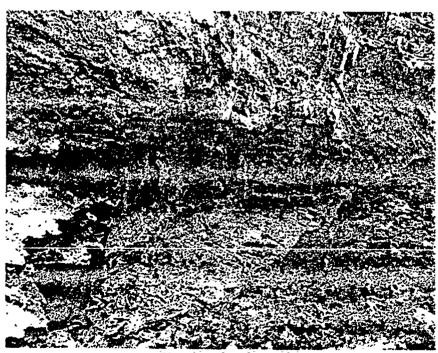


Photo. No. 27

U/S half of the step face at Sta 25+80 \pm prior to placing concrete to eliminate the overhang.



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Photo. No. 28

Large open fracture at Sta 26+46 to 26+60 partially filled with grout. Openings below and to the left and right of the hard hat extend into the U/S face for an unknown distance.

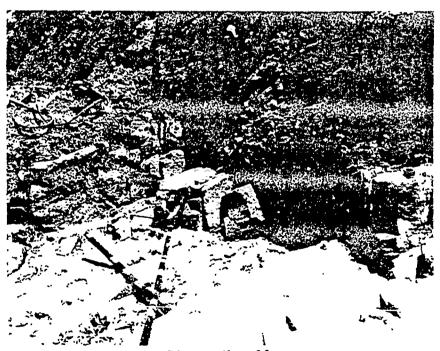
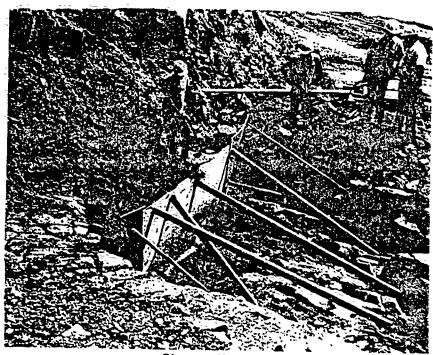


Photo. No. 29

Large open fracture at Sta 26440 to 26460 partially filled with grout. An opening to the right of the hard hat extends into the D/S face far an unknown distance. In the lower right-hand corner of the photograph a coke can plugs core Hole #37 at elev. 484. Note grout pipes in foreground to be installed in openings prior to backfilling with concrete.



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Photo, No. 30

Placing foundation protection concrete at Sta 26 ± 40 to 26 ± 60 the temporary form was used to aid in running concrete into the opening in the face.

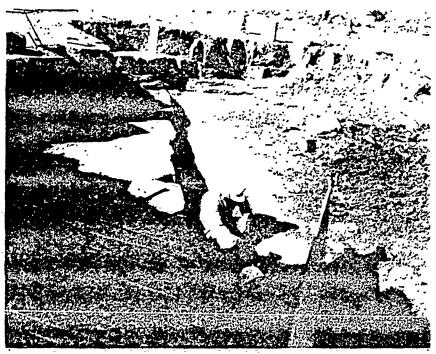


Photo. No. 31

A partially grout-filled fracture at Sta 26+40. This is the D/S extension of the large fracture at Sta 26+40 to 26+60. Clean up prior to backfilling with concrete.



Photo. No

This photograph shows the upstream extension of the fracture at Sta 26+40 to 26+60. Note the concrete (placed 7-2-73) plug in the packground.



This view (looking U/S from centerline) of the fracture at Sta 26+40 to 26+60 shows partial filling with grout at the lower right. This area was backfilled with concrete.

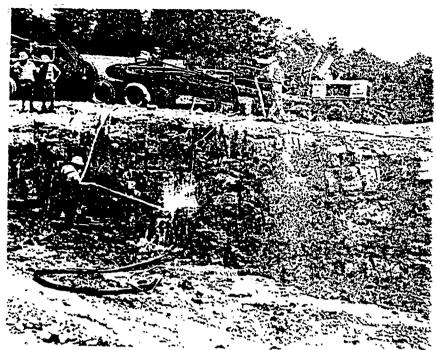


Photo. No 34

Clean-up prior to application of "Gunite" in left abutment. Note gunite machine mounted trailer.

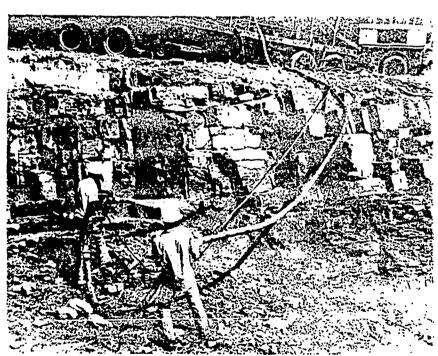


Photo No. 35

"Gunite" being applied to D/S face of C.C T in left abutment.

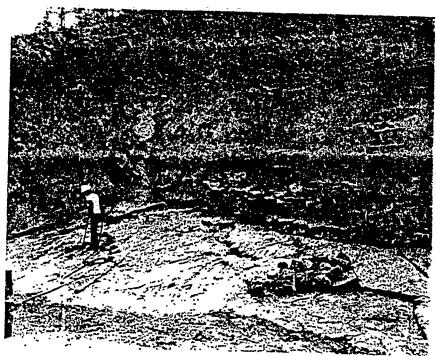


Photo. No. 36

"Gunited" D/S face and step face at Sta 26+90 to 27+10+.

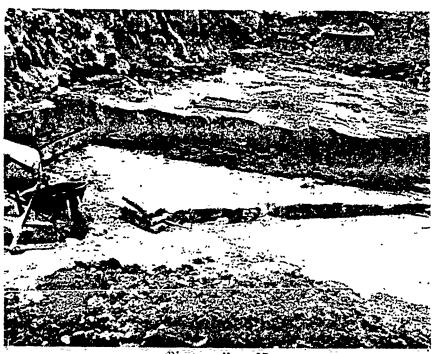


Photo. No. 37

A small portable mixer was used to produce concrete to eliminate small overhangs. This photo was taken at Sta 26 \pm 40 \pm D/S of centerline.



Photo. No. 38

Sand grout is being mixed by hand to back I in the first of tending D/S from the previously concreted area at 1.-1 m of the first lies under the broom at the right edge of the first sectors.

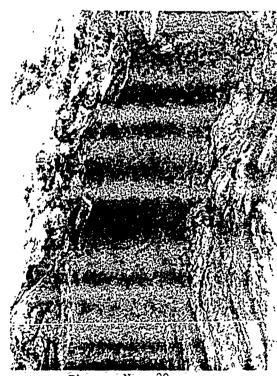


Photo. No. 39 A view within the right abut. of the fracture 35' U/S of centerline.

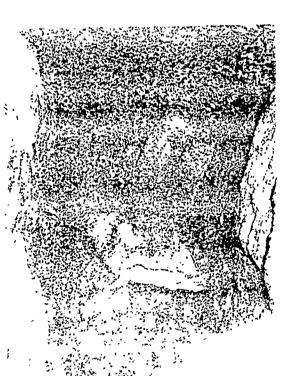


Photo. No. 40

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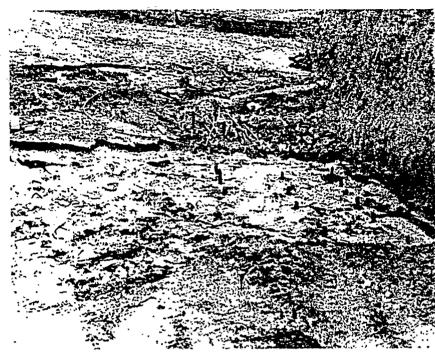


Photo. No. 41

Foundation along centerline at Sta. 25+50 to 25+70. Note three lines of grout holes in this area.



Photo. No. 42

Foundation from Sta 25+50 to 25+70 as seen from a point 40 feet D/S. The open fracture at left is to be backfilled with concrete.

APPENDIX C TEST EMBANKMENT REPORT TULSA DISTRICT

GILLHAM LAKE
COSSATOT RIVER, ARKANSAS
PERIODIC INSPECTION AND CONTINUING EVALUATION
OF
COMPLETED CIVIL WORKS STRUCTURES

TEST EMBANKMENT REPORT
GILLHAM DAM
TULSA DISTRICT

EXHIBIT A

(APPENDIX II)

TEST EMBANKMENT REPORT Gillham Dam, Tulsa District

- 1. <u>General</u>. The purpose of the test fill program was to develop engineering data to be used in preparing plans and specifications for construction of Gillham Dam, Cossatot River, Arkansas. The primary objectives were as follows:
 - a. Size and gradation of rock-fill material obtainable from the test quarry by varying blast-hole patterns and dimensions and by varying quantity and types of explosives.
 - b. Compaction of weathered and unweathered rock-fill materials by two different types of rollers (vibratory and pneumatic) using lifts of varying thickness.
 - c. Breakage and/or degradation of the rock-fill material due to rolling.

The test fill program consisted of one large rock-fill section having seven test panels or lanes (Nos. 1, 2, 3, 3A, 4, 5, and 6). Construction began on 19 June 1964 and was completed on 7 August 1964. Supervision for the project was furnished by the Construction Division, with technical assistance provided by the Engineering Division, Tulsa District. The information contained herein was largely taken from the Test Embankment Report, Gillham Dam, August 1964, and supplementary memoranda furnished by the Tulsa District.

2. Rock Type. The Gillham test fill consisted of two types of rock:
weathered sandstone and fresh sandstone. The weathered sandstone,
distinguishable by its iron-stain color was hard, dense, quartzitic, finegrained, and slightly fractured. The weathering consisted of leaching
and iron staining concentrated along bedding planes and joints. Blast
fractures across the bedding and joint patterns revealed fresh blue-gray

sandstone that was very hard and dense in the center of individual blocks. After blasting, the weathered material had angular particle shapes with about 10 percent of the rock degraded to sand sizes. The fresh or unweathered sandstone was very hard, brittle, and blue-gray in color. It appeared to be mildly metamorphosed with characteristics of quartzite. Scattered hairline fractures up to 1/8 in. wide were noted throughout this material. All of these fractures were tight to well healed with secondary quartz mineralization. Maximum rock sizes used in the test fill are given in table 1.

3. Description of test fill.

- a. The test fill was constructed on a leveled area approximately 1600 ft northwest of the test quarry site. Preceding fill placement, about 4 ft of badly weathered shale was removed to expose a firm shale suitable as a foundation. Settlement plates were not installed at the foundation level.
- b. The test fill consisted of seven sections or panels, as shown in fig. 1. Each panel was 18 by 50 ft, with 24- to 27-ft-wide (at completion) transition zones between panels. A IV on 10H slope, serving as a ramp for the construction and compaction equipment, was provided on either side of the embankment. Panel 3A was added on the end of the fill (see fig. 1) after construction was underway to supplement erratic data obtained from panel 3. Rock type, maximum rock size, loose lift thickness, number of lifts, compaction equipment, and number of passes for the seven test panels are given in table 1.

4. Construction.

a. Descriptions of excavation, hauling, and compaction equipment

are given in table 2. Rear-dump hauling units, loaded at the test quarry by power shovel, transported the rock directly to the test fill (no stock-piling was allowed) and dumped it at the leading edge of the advancing loose lift. Spreading to the desired loose lift thickness was then accomplished by a D-8 bulldozer. Oversize rock was controlled in the quarry by selective loading. Any oversize material reaching the test embankment was wasted during spreading. The shale content of the material was kept to a minimum by selective removal where possible. However, much interbedding of shale in the sandstone was encountered, and selective removal was not always feesible. Material for the last three lifts of panel 5 and all of panel 6 was passed through a screen to remove the minus 3-in. fraction. All other material was quarry-run.

b. After spreading, each lift was smoothed by one pass of the Bros Monel VP-20D 10-ton vibratory roller with the vibratory unit off or by one pass of the Ferguson 50-ton pneumatic roller, depending upon the specified method of compaction for that lift. A 6-ft grid pattern was then laid out on the smoothed lift surface from reference points beyond the test zone limits and subsequently marked with spray paint for easy identification. Initial level readings were taken on these points to establish the loose lift thickness (by comparing the new readings with the final readings on the underlying foundation or compacted lift). Compaction was then begun using either the vibratory or pneumatic roller, both towed by a D-7 bulldozer at speeds of 1 to 1-1/2 mph. The roller was towed over the test panels in alternate directions. Each lift was subjected to a total of eight passes by the roller, with level readings to measure surflement taken after every two passes.

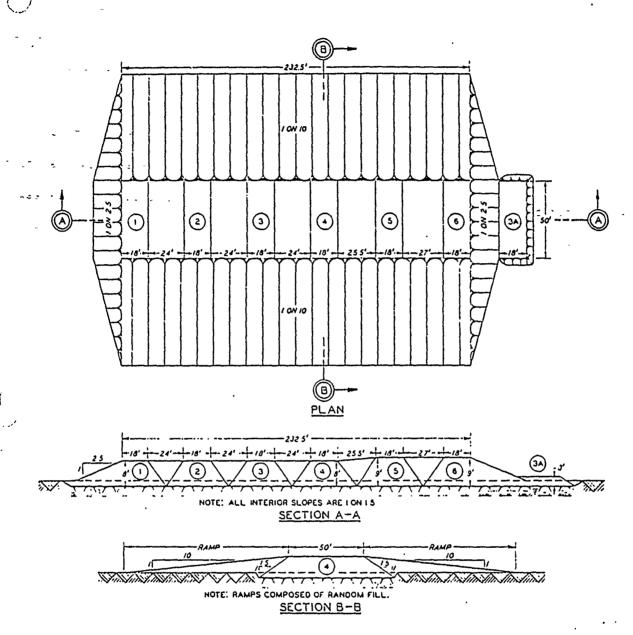


Fig. 1. Gillham test fill layout

Construction Details, Gillham Test Fill Table 1

No of Dance	ı	ry 8	ry 8	atic 8 .	atic 8	atic 8	ory ' 8	ory 8
	Roller	Vibratory	Vibratory	Pneumatic	Pneumatic	Pneumatic	Vibratory '	Vibratory
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Lift	Thickness in.	12	24	12	12	24	18	36
Maximum Rock Size Approximate	Spherical Diameter* in.	11-1/5	17	11-1/2	11-1/2	17		18
Ma	Weight 1b	80	250	80	80	250	200	300
	Panel Material	Weathered	Weathered sandstone	Weathered sandstone	3A Weathered sandstone	Weathered sandstone	Fresh sandstone**	Fresh sandstone (plus 3-in. only)
	Panel	1	8	m	3A	4	ស	9

^{*} Based on a specific gravity of 2.61 ** Last three lifts had all minus 3-in. material removed

Table 2

Construction Equipment, Gillham Test Fill

Item	Function	Description
Power shovel	Loading	Lima Model 604, 1-1/2-cu yd bucket capacity
Trucks (3)	Hauling	Euclid Model R-15, rear dump (2) Euclid Model 91-FD, rear dump (1)
Tractor	Spreading	Caterpillar D-8 with dozer blade
Tractor	Towing	Caterpillar D-7 with dozer blade
Compactors (2)	Compacting	Bros Model VP-20D, 10-ton, vibratory Ferguson 50-ton pneumatic
Separator	Rock separation	3-in. screen with a wobble-type feeder adjusted to 3-1/4-in. openings and equipped with a belt-loader

5. <u>Tests and measurements</u>.

a. Procedures.

- (1) Settlement measurements. Measurements of settlement were the primary means of assessing compactive effort. The grid marked on the surface of each lift to delineate the 24 points of settlement measurement is shown in fig. 2. Initial level readings were taken as discussed above. An average elevation at each point of the grid was obtained by placing the level rod in the center of a 12- by 12- by 1/2 in. steel plate positioned over the grid point.
- (2) Density tests. Field density tests were made at the conclusion of construction in the top lifts of panels 3, 3A, 4, 5, and 6. The tests were performed by excavating through the top lift of the panel from within a 6- by 6-ft wooden guide frame placed on the surface. Excavation of the hole was by hand labor, with each of the larger rocks being weighed individually and the smaller particles in groups. An extensive number of tape measurements were made of the finished pit and averaged in an effort to obtain accurate dimensions from which to compute the volume of the hole.
- (3) Mechanical analyses. Gradation tests were run on the material excavated from the density test pits in panels 3, 3A, 4, 5, and 6. These analyses established the after-compaction gradations for the top lifts of those five panels. These tests were run by first weighing the total sample, then grouping the rocks into weight ranges, and finally computing the percent of the total sample represented by each weight range. The percent smaller than 3 in. was determined by actually sieving the finer fraction over a 3-in. sieve.
- (4) Inspection trench. After completion of all panels of the test fill, an inspection trench was excavated with the D-8 bull-dozer. The trench, which had a base width of 14 ft, exposed all lifts of all panels in the embankment. The inspection trench permitted visual assessment of the compaction characteristics of the fill.

b. Results

(1) Settlement measurements. Figures 3 through 7 show the percent settlement of each lift of panels 1, 2, 3, 3A, and 4 which were constructed of quarry-run weathered rock. The average settlement of each of these panels is shown in the respective figures and in the plot of fig. 8. Figures 9 and 10 present the lift settlement data for panels 5 and 6, which were composed of plus 3-in. grizzled fresh rock except for

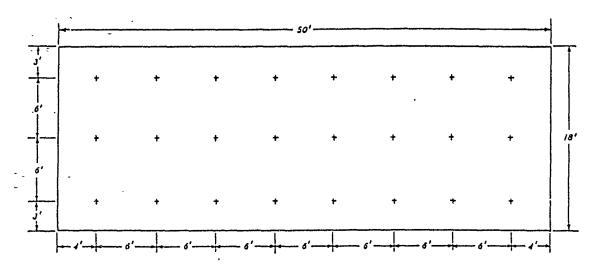


Fig. 2. Grid layout for level readings, Gillham test fill

lifts 1, 2, and 3 of panel 5, which contained quarry-run fresh rock. The average settlement curves for panels 5 and 6 are also shown in figs. 9 and 10 and plotted separately in fig. 11. The immediately noticeable aspect of the settlement data for the panels constructed of weathered rock (panels 1. 2. 3. 3A, and 4) is the erratic data for several of the lifts. The measurement data for panels compacted with the 50-ton pneumatic roller (3, 3A, and 4) were less consistent than those for panels rolled with the vibratory unit, which might be expected because of the much smoother surface left by the vibratory roller. Further examination reveals that the most variable results were for the 12-in. lifts, regardless. of the compaction equipment used. The erratic nature of the settlement data from the 12-in. lifts of panels 1, 3, and 3A probably resulted from erroneous settlement readings or the presence of oversize rock. The data (see fig. 8) indicate, however, that slightly better compaction was obtained with 12-in. lifts of weathered sandstone, and that four passes of either type roller produced most of the settlement achieved by eight passes for this lift thickness. The settlement plots for panels 5 and 6 (fresh sandstone), shown in figs. 9 and 10, are much less variable than those for the other panels. The data in figs. 9 and 10 indicate that the plus 3-in. material has superior compaction characteristics than did the quarry-run material. Comparison of figs. 9 and 10 indicates that the use of 18-in. lifts resulted in more efficient compaction than did the use of 36-in. lifts. Figure 10 shows that the rate of settlement for the 36-in. material had not decreased even after eight passes of the roller, and only 9 percent settlement was attained at that point. By comparison, fig. 9 shows that the same material in 18-in. lifts reached 14 percent settlement after six passes with a marked decrease in the rate of settlement beyond six passes.

- (2) Density tests. The results of density tests made in panels 3, 3A, 4, 5, and 6 are summarized in table 3. It was observed during the density tests taken in panels 5 and 6 that the rock-to-rock contact produced by the vibratory roller resulted in an unusually high degree of stability. Considerable pick work was required to loosen the rock for excavation. The vertical sidewalls of the pits had no tendency to slide or slough, but were tight and stable.
- (3) Mechanical analyses. After-compaction gradation tests were performed on the material taker from the density tests in panels 3, 3A, 4, 5, and 6. The after-compaction curves for material in panels 3, 3A, and 4 are shown in fig. 12. Since there were no before-compaction data for these panels, no assessment of particle breakage can be made. The gradation curves resulting from tests in panels 5 and 6 are given in figs. 13 and 14, respectively. Also shown in figs. 13 and

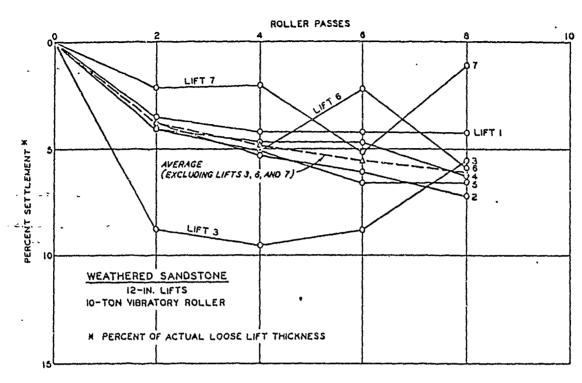


Fig. 3. Percent settlement vs roller passes, panel 1, Gillham test fill

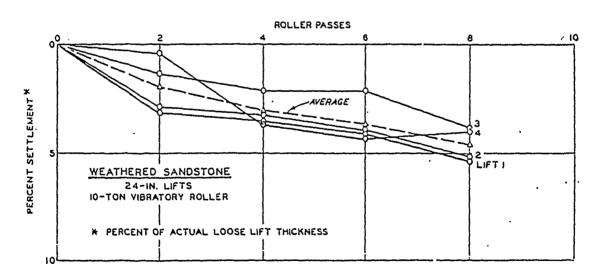


Fig.4 . Percent settlement vs roller passes, panel 2, Gillham test fill

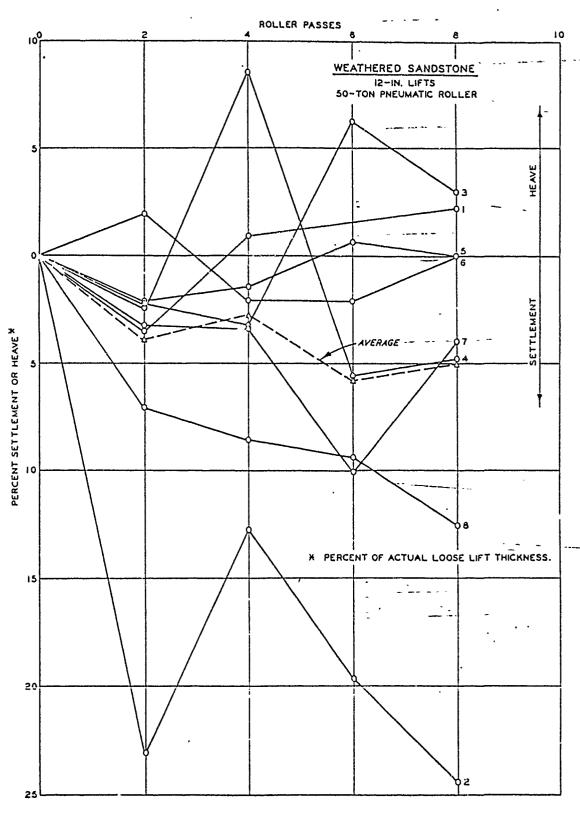


Fig. 5 . Percent settlement vs roller passes, panel 3, Gillham test fill

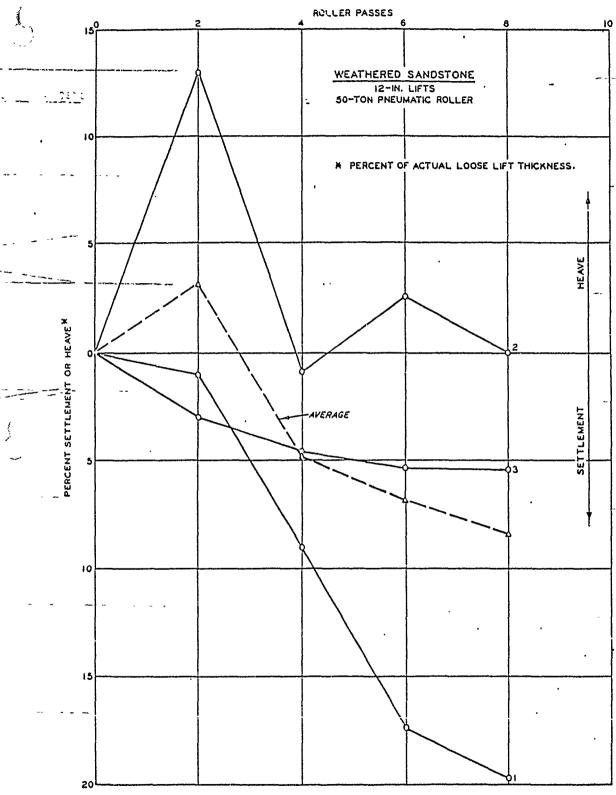


Fig. 6 . Percent settlement vs roller passes, panel 3A, Gillham test fill

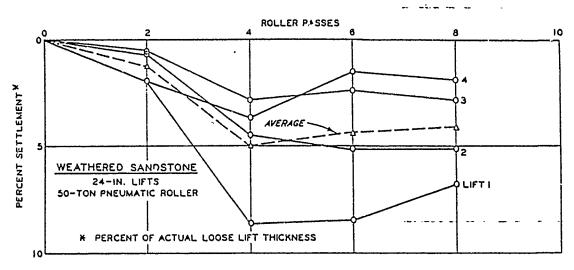


Fig. 7. Percent settlement vs roller passes, panel 4, Gillham test fill

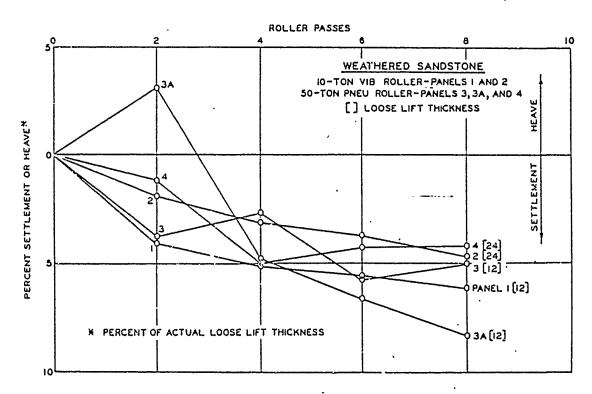
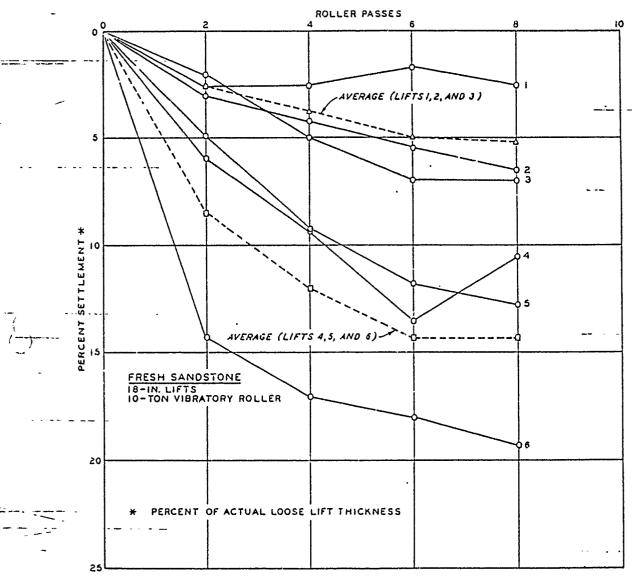


Fig. 8. Average percent settlement vs roller passes, panels 1, 2, 3, 3A, and 4, Gillham test fill



NOTE: LIFTS 1,2, AND 3 CONSTRUCTED OF QUARRY-RUN FRESH ROCK. LIFTS 4,5, AND 6 CONSTRUCTED OF PLUS 3-IN. FRESH ROCK.

Fig. 9. Percent settlement vs roller passes, panel 5, Gillham test fill

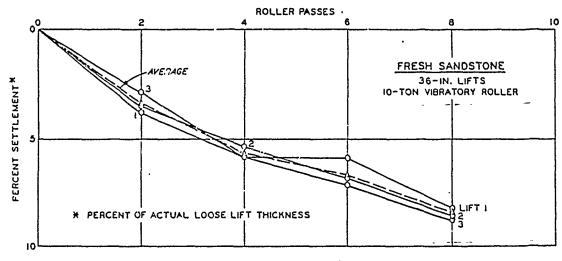


Fig. 10. Percent settlement vs roller passes, panel 6, Gillham test fill

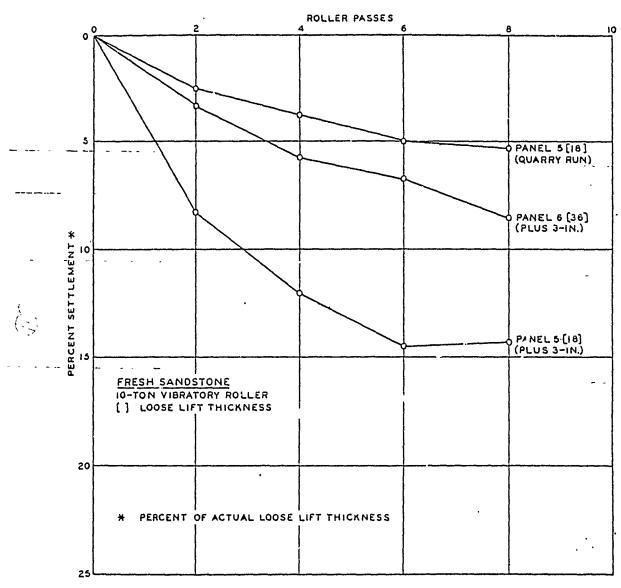


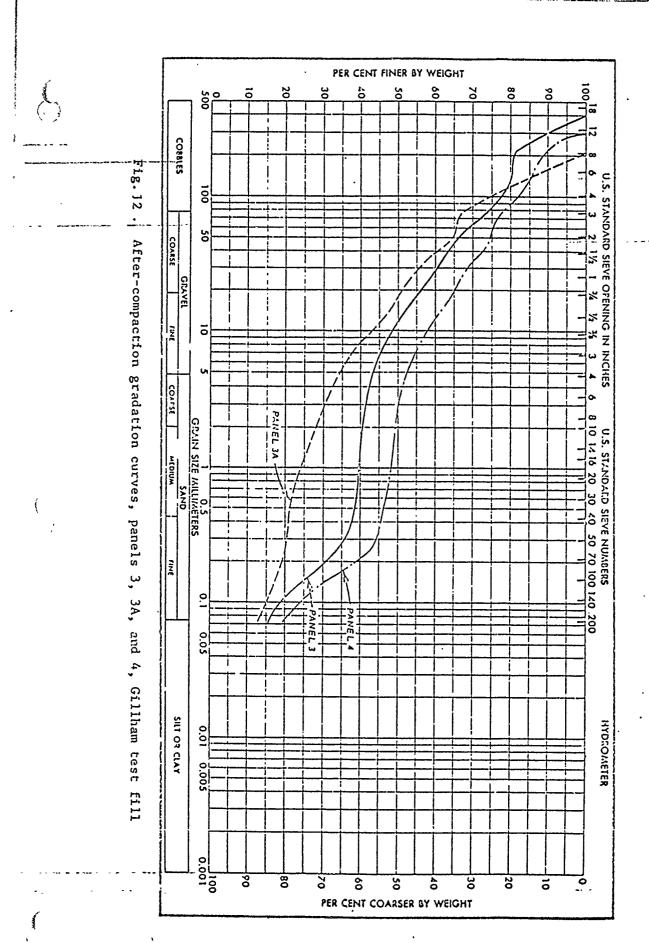
Fig. 11. Average percent settlement vs roller passes panels 5 and 6, Gillham test fill

Table3

Results of Density Tests After Compaction

AMARILA STATE STATES	Panel No.	Lift Thickness in.	Material*	Roller	In Situ Density pcf	Porosity
	3	12	Weathered sandstone	Pneumatic	135	17
	3A	12	Weathered sandstone	Pneumatic	128	21
	4	24	Weathered sandstone	Pneumatic	137	16
	5	18	Fresh sandstone	Vibratory	91	43
	. 5	18	Fresh sandstone	Vibratory	83	48
	6	36	Fresh sandstone	Vibratory	104	36

^{*} All weathered sandstone was quarry-run; all fresh sandstone was plus 3-in. material.



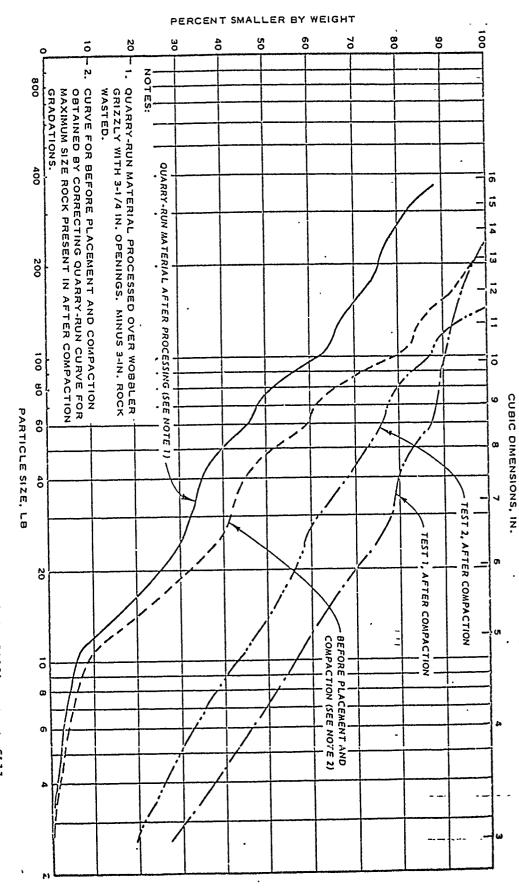


Fig. 13. Before- and after-compaction gradation curves, panel 5, Gillham test fill

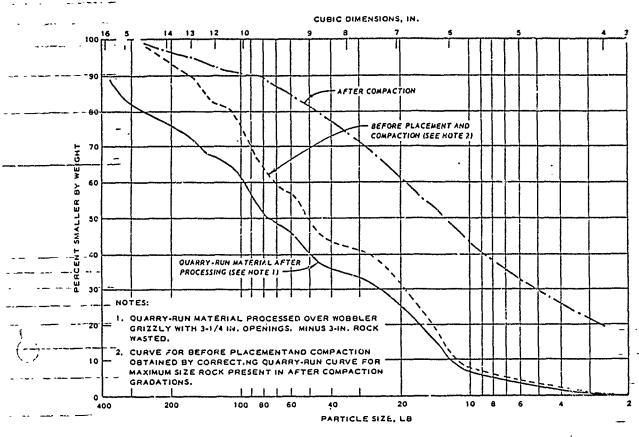


Fig. 14. Before- and after-compaction gradation curves, panel 6, Gillham test fill

14 is the curve for the quarry-run fresh rock after processing over a wobbler grizzly with 3-1/4-in. screen openings. The curves representing the material before placement and compaction were derived from the processed quarry-run data by correcting to the maximum size rock present in the after-compaction gradations. It is seen from figs. 13 and 14 that significant degradation occurred under the action-of the vibratory roller. Field personnel noted that 85 percent or more of the larger size rock chunks were broken at least once by the vibratory compactor.

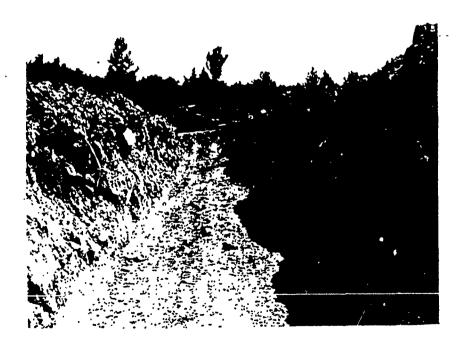
(4) Inspection trench. Visua' examination of the inspection trench confirmed the tightness and unusually high degree of stability of all compacted material. No mention was made of any segregated or stratified zones or that any evidence of poor bonding between lifts was found. Two-views of the inspection trench are shown in the photographs in fig. 15.

6. Discussion.

- a. As mentioned earlier, the settlement data indicated more settlement was obtained using the plus 3-in. fraction rather than the entire (quarry run) portion of the fresh sandstone. This also occurred at the Laurel Dam test fill, and, although no mention was made by the district about stratification and excess surface breakage, it is probable that the cause was not only that there were more open voids in the plus 3-in. material to begin with, but that a dense layer of surface fines formed in the quarry-run material, which reduced the compaction obtained in the lower part of the lift.
 - b. The results of the density tests on the fresh sandstone fill material (91, 83, and 104 pcf) seem low, especially in view of both the description given of the density pit sidewalls (see para 5b(2) and the settlement data. All things considered, one would tend to suspect the accuracy of the density tests. Large-volume density tests are themselves crude, even when the best possible procedures are used. In these tests,



a. View north, inspection trench excavation



b. View south, inspection trench excavation

Fig. 15 Inspection trench, Gillham test fill

pit volume was determined by geometric measurements which is not desirable for a rock-fill material. Computations of volume made from rule or tape measurements usually fail to account for protruding rocks, cavities, and unsymmetrical pit shapes. It should be recognized, however, that it is possible to obtain a very stable, free-draining, and otherwise desirable rock fill without necessarily achieving a correspondingly high density.

c. The erratic nature of some of the settlement data indicates the problems involved in taking level readings on a rock-fill surface. Often a premarked point for a reading will consist of rock that has rotated under the roller action rather than having been pushed down, and thus does not reflect the average elevation of the surrounding surface. This is often true in the case of compaction by rubber-tired rollers because they do not tend to crush the surface rock as does a vibratory roller. This problem might also be aggravated by the use of thinner lifts. Some inaccuracies could also be caused if significant settlement is still occurring in underlying lifts but is being measured as occurring in the lift under compaction. An alternative to this problem would be to plot the data as actual settlement for an entire panel (as long as all lifts in the panel were of the same thickness and material). However, if settlement by lift in terms of percent of initial lift thickness is desired, the preceding lift can be rolled until no significant additional settlement is observed.

APPENDIX D

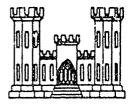
SWDGL REPORT NO. 7491

RESULTS OF TESTS OF BORROW MATERIALS

SWDGL REPORT NO. 7491

RESULTS OF TESTS OF BORROW MATERIALS

GILHAM DAM - TULSA DISTRICT



CORPS OF ENGINEERS

U. S. ARMY

SOUTHWESTERN DIVISION LABORATORY

DALLAS, TEXAS

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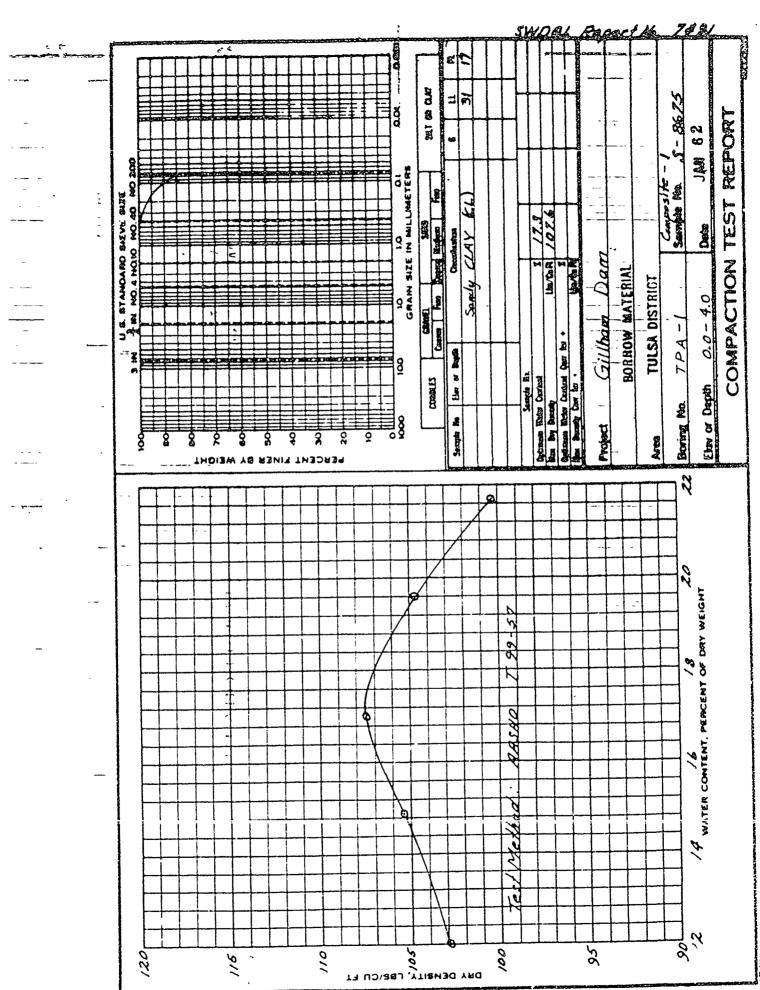


PLATE NO. 2

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	The State of the S	S 10 K M				Sample No. 5-8475 Dete JAN 62 SSION TEST REPORT
eres (1970)	THEAR STRESS, TONS/SO FT	O S APPLIED PRINCIPAL	Remarks: Q test		Project Gillham Dam BORROW MATERIAL Area TULSA DISTRICT	M No.
	T DEVIATOR STRESS, TONS'SO FT	BYIN BEBCEN	VOLUMETRIC ST	→ × :	Constant Strain, 003 III Control Control Un Consolidated, Un Drained	4.8° Ian Ø fration Sandy
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	TO STANDARD U.S. STA	Test No Water Content, Wo Z Dry Density Lbs/Cu Ft	Saturation, So X W.C. affer Saturation, W, X Saturation, S X	Consol Pressure 1/SqFt W.C. after Consol, W.c. & Vool Ratio after Consol e.	Min Prin Stress, cr. 1/Sq Ft Min Prin Stress, cm. 1/Sq Ft Water Content, W, Z Void Ratio, e,	Specimen Diameter Inches Initial Height Inc

PLATE NO. 4

	PATTER TO THE PA	· · .	I REPORT NO. 7481
1		8	K-1 Me S-8675 (95% f, ω) JAN 62 TEST REPORT
	11-08 2NOT 2E3RIZ RA3HE 5 5 7 7 4 7	BERECHT. R Test	Propect Gilham Dam BORROW MATERIAL Area TULSA DISTRICT Boring No TPA-1 Elev or Depth O.O4.0 TRIAXIAL COMPRESSION
		VOLUMETRIC STRAIN PE	Type of Text - R Constant Strain, 0.03 /min Control XConcated Strain, 0.03 /min Control XConcated Strain, 0.03 /min Control XConcated Strain, 0.03 /min Control Inpe of Spacemen REMOLDED Ø - 27.6° Im Ø - 522 c-0.21/Sq ft Chashcaten Sondy CLAY (CL) LL 3/ G 2.66
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Type of Specimen REMOLDED Classification Soldy CLAY (CL) LL 31 G 2.66 PL 17 Day

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APPENDIX E

SWDGL REPORT NO. 8458

RESULTS OF TESTS OF BORROW SOIL

SWDGL REPORT NO. 8458

RESULTS OF TESTS OF BORROW SOIL

GILLHAM DAM - TULSA DISTRICT



CORPS OF ENGINEERS
U. S. ARMY

SOUTHWESTERN DIVISION LABORATORY

DALLAS, TEXAS

,		REQUEST FOR AN	D RESULTS	OF TESTS		PAGE		
-		FOR TEST 17	11-14-63-23	dated 13 May	1962			
	1. TO:	. Name of the contract of the		2. FROM:			7.1903	
	Southwestern Division Laboratory US Army Engineer Division, Southwestern Corps of Engineers				, Fam Branc my Engineer	District,	ulsa	
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•	3. PRIME CONTRACTOR AND ADDRESS			4. MANUFA	CTURING PLANT NA	ME AND ADDRESS		
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	5. END ITEM AND/OR PE		6. SAMPLE	P. O. NUA 7. LOT NO.		SUBMITTAL	9. DATE	
	Gillham Dem		NUMBER				SUBMITTED	
	10. MATERIAL TO BE	10s. QUANTITY SUBMITTED	11. QUANTI	ſΥ	12. SPEC. & AME	NO AND/OR DRAWING	NO. & REY.	
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	13. PURCHASED FROM OR	Ten Bags	i4. SHIPMEI	T METHOD	15. DATE SAMPL	ED AND SURMITTED BY		
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	Holes TPB-8	ECIAL INSTRUCTIONS AND/OR WAI		Vehicle				
	17.	•	960	EIPT				
	THE ABOVE MATERIAL HAS	BEEN RECEIVED				. — <u></u>		
	DATE	TYPED NAME AND TITLE			SIGNATURE			
	9 May 63	A. H. FEESE, Engin	eer	İ				
	3. RESULTS OF TEST (Continue on plain white paper if more space is required)							
	1. RESULTS	MOICATED		2. DATE REC'D	J. DATE REPORTED	4. LAB REPORT NUMBER		
	OTHER (Specify)				9 May 63	29 July 63	·	
	5. TEST PERFORMED, RESULTS OF TEST SAMPLE RESULT REQUIPEMENTS							
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		This report	complete	es reques	sted testing	g.		
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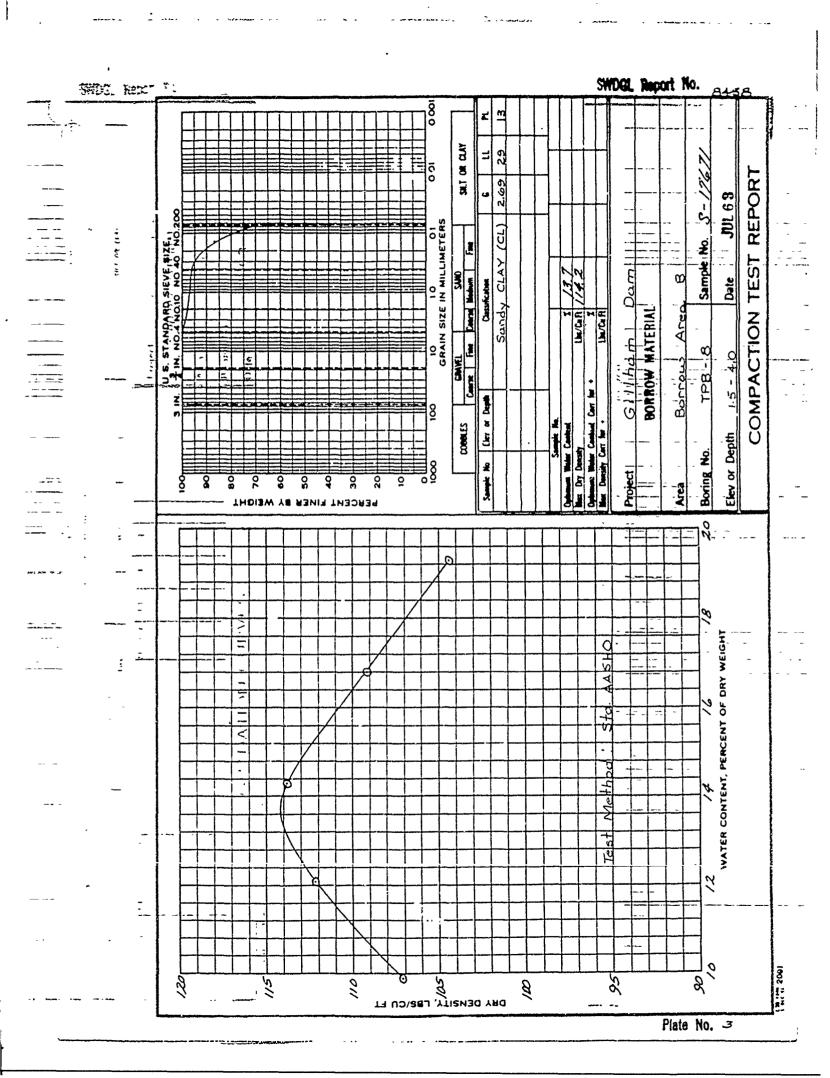
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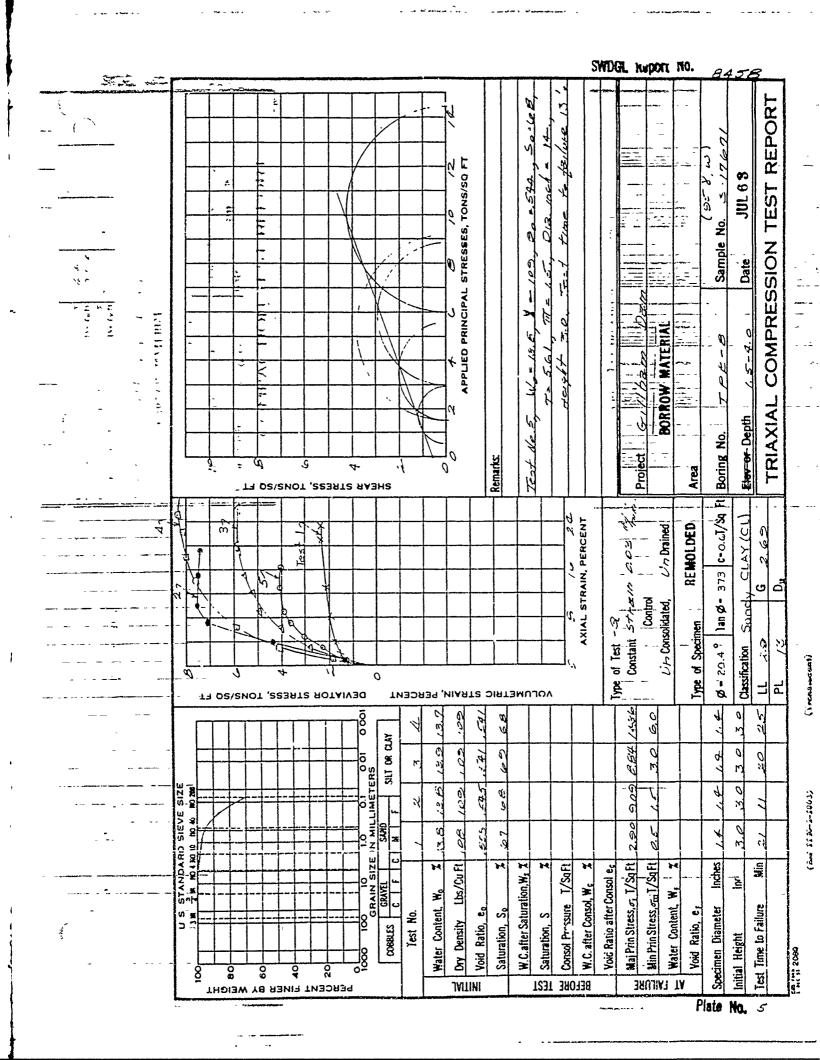
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Sill II SMDGL Report No. 8458 0000 SALT OR CLAY # COMPACTION TEST REPORT 2.69 10 10 01 CRAIN SIZE IN MILLIMETERS g 1212131884 ... BORROW MATERIAL 2.0 - 6.0 GRAVEL TPC - 14 Ser 04 COBBLES Elev or Depth Boring No. ō PERCENT FINER BY WEIGHT 22 ASHO DRY WEIGHT WATER CONTENT, PERCENT OF DRY DENSITY, LBS/CU FT 25 110 100 Plate No. 4



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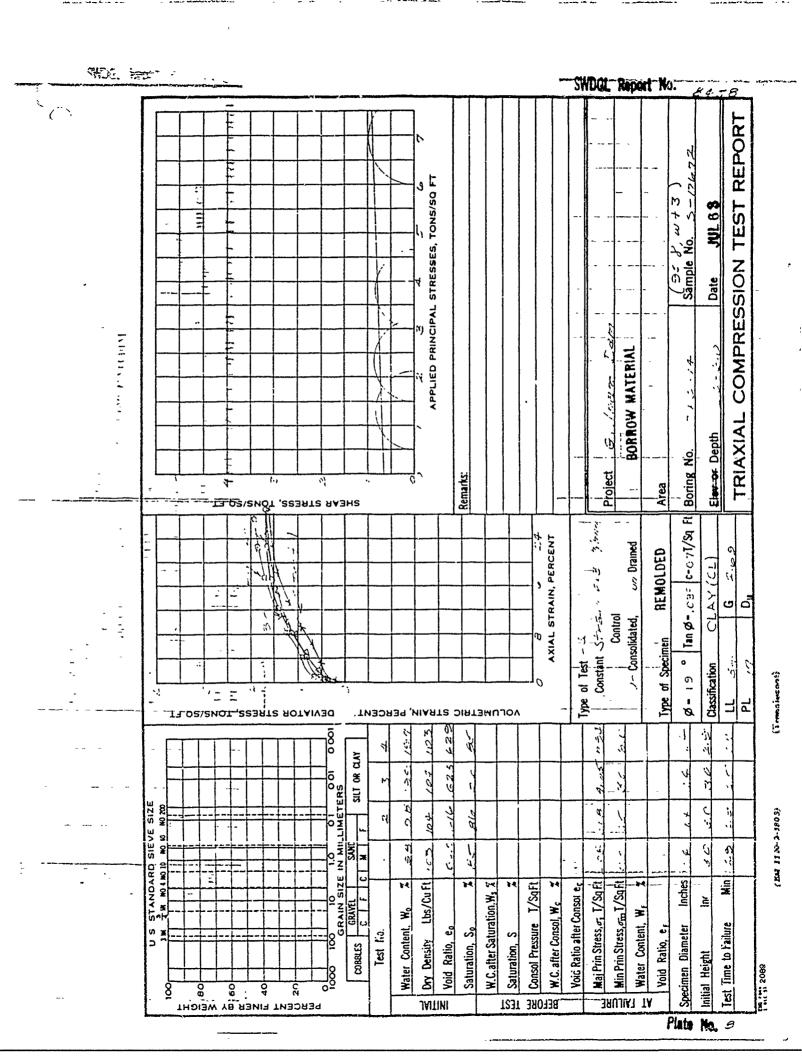
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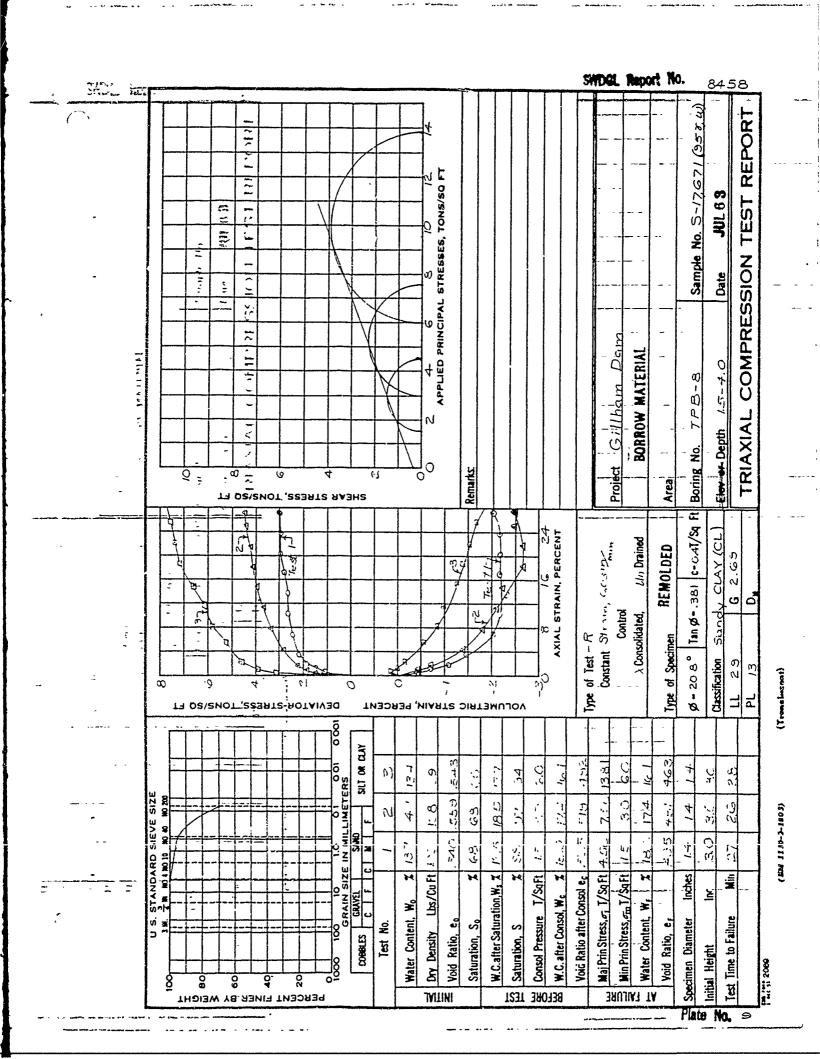
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PERCENT FINER BY WEIGHT

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	,	SWDGL F	eport	No.	84	3 8	
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APPENDIX F

SWDGL REPORT NO. 8473

RESULTS OF TESTS OF ROCK,

ROCK FILL, RIPRAP AND CONCRETE AGGREGATE

SWDGL REPORT NO. 8473

RESULTS OF TESTS OF ROCK

ROCK FILL, RIPRAP AND CONCRETE AGGREGATE

GILHAM DAM - TULSA DISTRICT



CORPS OF ENGINEERS

U. S. ARMY

SOUTHWESTERN DIVISION LABORATORY

DALLAS, TEXAS

3 10 10 10 11 11 11 11 11 11	A REQUEST FOR TEST TIME. Southwestern Bivision Laboratory US Army Engineer Division, Seathwestern Corps of Engineers 4815 Case Street Dallas, Turns 75235 2 PRIME CONTRACTOR AND ADDRESS 4. MANUFACTURING CONTRACT NUMBER 5. END ITEM AND/OR PROJECT CONTRACT NUMBER 6. SAMPLE FOR NUMBER 7. LOT NO 8. R NUMBER 9. LOT	RING PLANT NAME REASON FOR SU THE TOP THE 2. SPEC. 2 AMEND A FOR SAMPLE & D.	BMITTAL BMI	9 DATE SUBMITTE								
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and the same

Missing Protect Track Results of Tests of Quarry Rock, Investigation for Mill Material Gillham Dom - Tules District

- 1. Reference Test Request No. TU-FM-63-22 deted 13 May 1963 from the Chief, Foundations and Materials Branch, Tulsa District.
 - 2. The following materials were received 9 May 1963:

	SWD Sample No.	Material	Field Eals No.	Depth, feet	Quantity
	C-19765 A	2-1/8" Core, Sandstone	4 6	88.5 to 89.5	•
	В	#	Ħ	98.0 to 98.7	•
	C -	**	92	102.1" to 103.1	•
	D	•	11	152.0 to 153.0	•
	K	м	Q - 7	34.6 to 95.6	•
- A second decimal of the	F	N	*	131.4 to 132.8	•
	G	99	76	135.3 to 136.1	•
	н	11	*	150,0 to 150.8	•
	I	#	11	178.8 to 179.8	•
	J	n	Q - 3	95.9 to 96.6	•
	K	W	ĸ	105.2 to 106.6	•
	L	11	×	111.0 to 112.5	
	Ж	2-1/8" Core, Shale	9 - 7	104.9 to 105.4	•
™ w - —n was an ann	N	- M [*]	96	109.4 to 109.9*	•
	C-13766	2-1/8" Core, Shale	Q - 10	10.0 to 20.0	75 lbs
	C-19767	Quarry Sandatone	œ	SSE CONTRACTOR OF THE CONTRACT	4000 lbs

*Test request listed 108.4 - 109.4. The sample received was marked as shown.

3. The requested tests have been completed and results summarized as follows:

Results of Physical Tests of 2-1/8" Core

Table 1

Results of Tests of Riprap Material

Table 2 and Plate 1

4. The sample of quarry sandatone is being retained for additional tests as might be requested by Tulsa District.

The for Park Marketin.		QUARTZITIC SANDSTONE, gray, bard, fine grained. Contains occasional bair line shale seams and fractures well healed with calcite and quartz.	quarizific sandstone, ag adove.	QUARTZITIC SANISTONE, dark gray, fine grained, bard. QUARTZITIC SANISTONE, gray, fine grained, bard.	SHALE, black, with sandstone bands $1/2$ " to 1" thick at intervals of $1/2$ " - 3".	SHALE, SANDY SHALE end SANDSTONE mixed core	542	nevolutions with 12 balls. solutions with 12 balls. nours for each of 10 cycles.
TABLE NO. 1 (Revisea)	Tests of 2-1/8 lnca Rock Core Specimens re le Other Tests	Specific gravity SSD, 2.65	Absorption, U. 5%	Specific gravity SSD, 2.64 Absorption, 0.4%	•	Abrasion, L.A., \$ Loss(1) Representative sample as received - 41.3 All Residue from slaking tests - 39.2 Siaking Test, see note (2)		Test conducted on 16,000 gram sample of approximately 3" to 2" lengths of core subjected to 10000 Residue predominantly sandy soale., Representative pieces of the core were soaked in water 16 hours and exposed to laboratory air 8 No measurements were made but photographs before and after were taken and are shown on Plate 1.
Local Section 1991 And	Results of Test. Compressive	32300 28200 28200 45100 35500	34000 17400 25900 26000 20300	9400 15200 23800	7000	,	: :	<pre>ple of approximately 3" e.; s were soaked in water lé otographs before and afte</pre>
). Depth to Depth Feet	88.5 - 89.5 98.0 - 98.7 102.1 - 103.1 152.0 - 153.0	94.0 - 95.6 131.4 - 132.8 135.3 - 136.1 150.0 - 150.8 178.8 - 179.8	95.9 - 96.0 105.2 - 106.6 111.0 - 112.5	104.9 - 105.4 109.4 - 109.9	10.0 - 20.0	1	Test conducted on 16,000 gram sample of Residue predominantly sandy soale., Representative pieces of the core were s
	Field Hole No.	.o : : : : o'	C	Q, 1 = = Q,	5 - 2	Q - 10		Test conducte Residue predo Representativ No measuremen
SWDGL Report No. 8473	SWD Sample No.	. c-13705 A B C D	C-19765 E F G ' H	c-19765 J K L	C-19705 M	C-19766		NOTES: (1)

No measurements were made but photographs before and after were taken

Results of Tests of Sandstone Chunk Sample

SWD Sample No. C-19707

Tests	Results
Specific Gravity, Bulk SSD (crushed to 1" - 5/8") Unit Weight, lbs/cu.ft. (calculated from SSD sp. gr.) Absorption, \$ (crushed to 1" - 3/8") Soundness, \$ loss:	2.63 164.1 0.8
MgSO _i , 5 cycles, 2-1/2" - 1-1/2" fraction Freeze-thaw, 25 cycles, 2-1/2" - 1-1/2" fraction Abrasion, L.A., \$ loss	1.1
"A" grading "E" grading	22.1 15.4

Description: This sample of ledge rock consisted of quartzitic SAMESTONE, gray, hard, dense, fine-grained, weathered and slightly fractured. The weathering consisted of leaching and ferruginous staining to a depth of 1/16" to 5", located on bedding planes and fractures. The staining appears to be in two distinct bands with the innermost being darker and marder due to it being a zone of iron concentration from the lighter leached band above it; this tends to have a slight case mardening effect on the rock. These bands appear on almost all chunks of any size and are of a uniform thickness for each chunk, although some irregularity was noted in the thicker bands. The fractures are mainline to 1/8", tight to well-healed with quartz. A small amount of talc-like material was noted on several broken fracture planes. Approximately 9 percent of the sample had chunks with lengths greater than three times their thickness. The maximum size chunk received was 2.0 x 1.2 x 1.2, and the average size was 1.1 x 0.7 x 0.6.



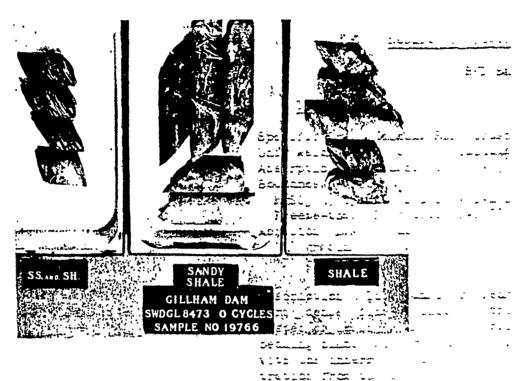
Photograph of Sample as Received

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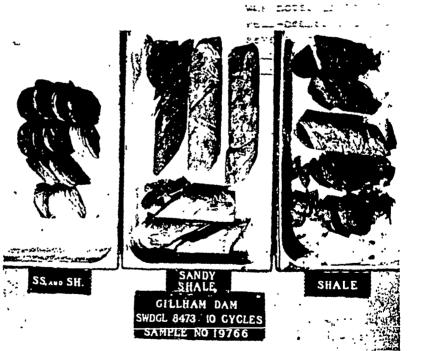
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COMPATIBLE SAMETOM STANDELLE THE CONTROL OF THE CON



Bufore Test



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APPENDIX G

PLATES

GILLHAM DAM AND RESERVOIR SALINE RIVER, ARKANSAS

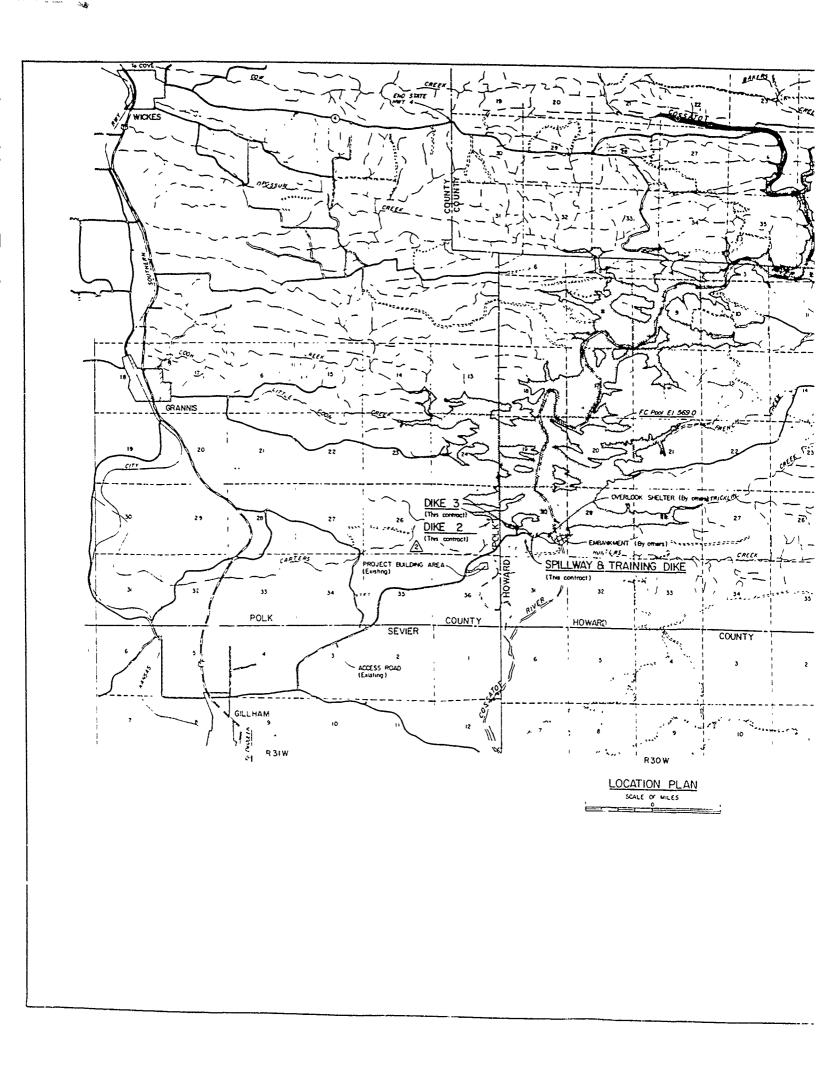
EMBANKMENT CRITERIA AND PERFORMANCE REPORT

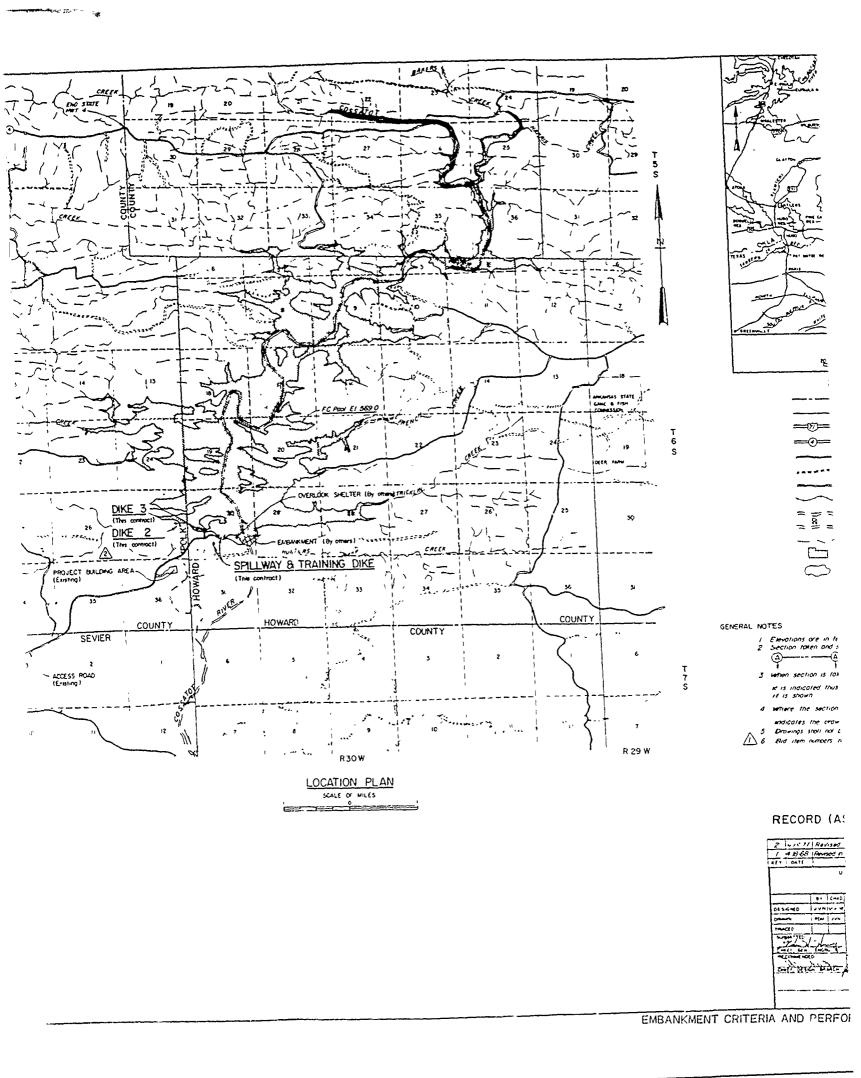
APPENDIX G - PLATES

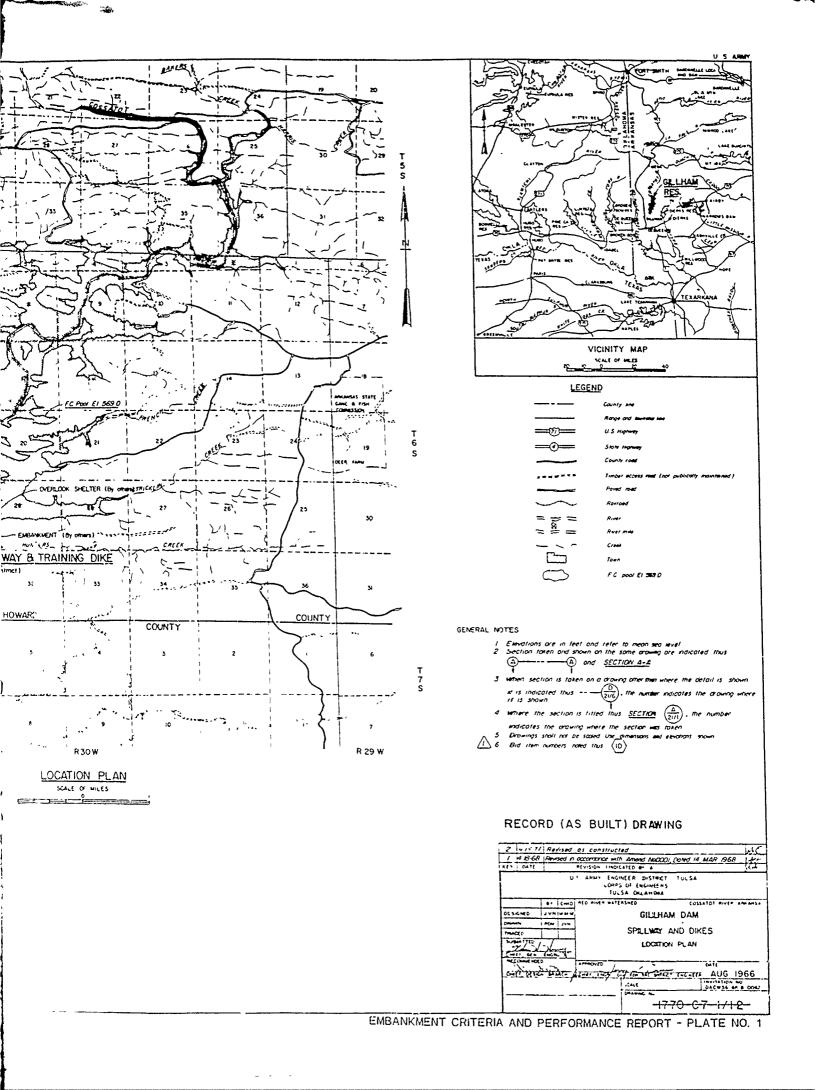
Plate No.	Drawing No.	<u>Title</u>
1	1770-07-1/1.2	Location Plan
2	1770-C7-1/1.2 1770-C7-2/1.1	General Plan and Sections
	•	
3 4	1770-C13-98/1.1	Embankment - Plan of Explorations and Log Section
4	1770-C7-98/1.1	Spillway and Dikes - Plan of Exploration and Sections
5	1770-C7-10/1.1	Spillway Excavation Plan
6		Foundation Report - Key Plan
7		Foundation Report - Spillway
8		Foundation Report - Left Chute Wall
9		Foundation Report - Right Chute Wall
10		Foundation Report - Chute Slab
11		Foundation Report - Section Spillway and Chute Slab
12		Foundation Report - Grouting Profile
13	1770-C5-98/1	Outlet Works - Plan of Borings and Log Sections
14	1770-DM9-8/2	Embankment - Diversion - Plan and Sections
15	1770-C13-12/1	Embankment - Plan, Profile and Sections
16	1770-C13-12/2.1	Embankment - Sections and Details I
17	1770-C13-8/2A	Embankment - Wire Fabric and Cable Installation
18	1770-C13-8/2	Embankment - Cable Anchorages
19	1770-C13-13/1	Embankment - Engineering Measurement Devices
20		Embankment Record Samples - "As Built" Shear Strengths
21	1770-DM9-98/7	Stability Analysis - End of Construction Upstream
22	1770-DM9-98/8	Stability Analysis - End of Construction Downstream
23	1770-DM9-98/9	Stability Analysis - Steady Seepage
24	1770-DM9-98/10	Stability Analysis - Sudden Drawdown
25	1770-DM9-98/11	Stability Analysis - End of Construction
26	1770-DM9-98/12	Stability Analysis - Steady Seepage
27	1770-C7-12/1.2	Spillway - Plan
28	1770-C7-12/2.2	Spillway - Embankment and Wraparound Sections
29	1770-C7-12/4.2	Dikes - Plan and Sections
30	1770-C7-21/2	Spillway - Typical Structural Arrangement
31	1770-07-21/3.3	Spillway - Typical Sections and Details
32	1770-C7-21/7.1	Spillway - Monolith 4 thru 8 - Weir Reinforcing
33	1770-C7-21/9	Spillway - Right Non-Overflow Monoliths 1 thru 4
34	1770-C7 21/20	Spillway - Left Non-Overflow Monoliths 8 thru 11
35	1770-C7-21/30.2	Spillway - Chute Slab
36	1770-DM12-21/4	Spillway - Overflow Stability Analysis
37		Spillway - Design Study
38	1770-DM12-21/5	Spillway - Non-Overflow Stability Analysis
39	1770-DM12-21/6	Spillway - Chute Wall Stability Analysis
40	1770-DM12-36/1.1	Spillway - Tainter Gates
		J

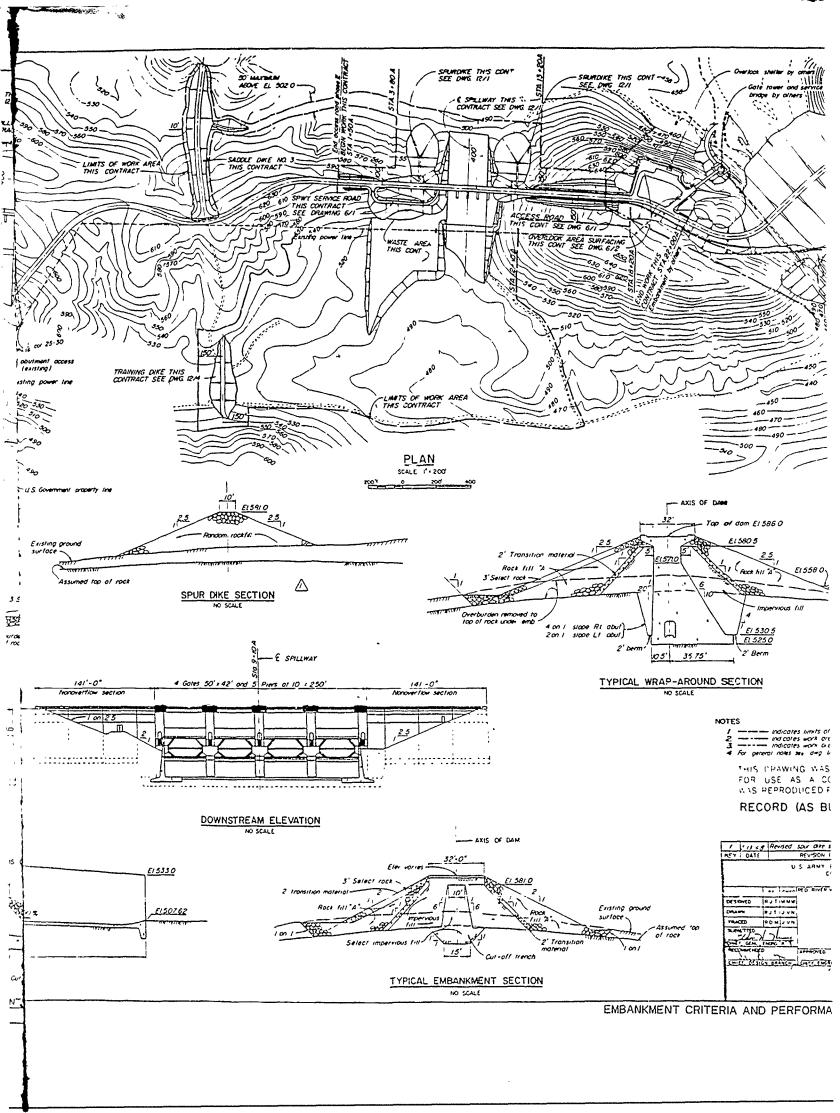
APPENDIX G - PLATES (continued)

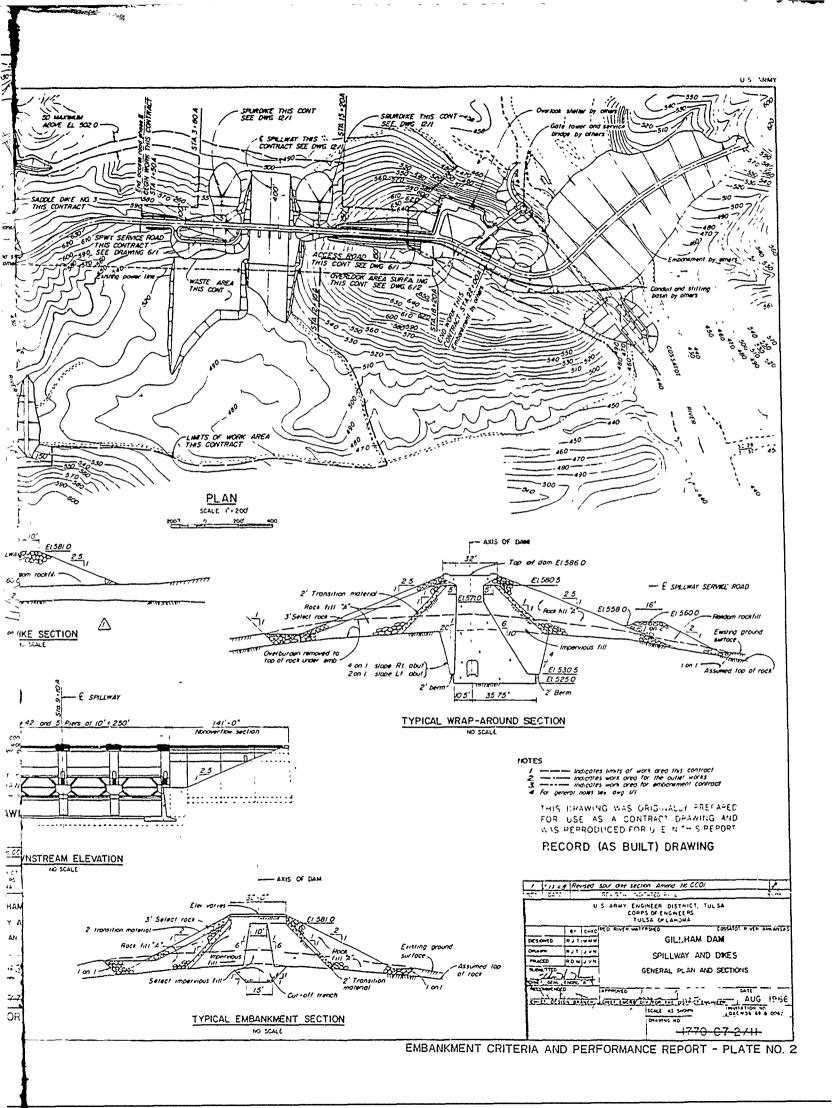
. 25717	Plate No.	Drawing No.	<u>Title</u>
	41	1770-C5-2/2.1	Outlet Works - Plan and Profile
	42	1770-C5-10/1.1	Foundation Excavation - Gate Tower and Inlet Channel
	43	1770-DM10-22/9	Low Flow Wet Well Facilities
	44	1770-C5-22/2.4	Gate Tower - Plan and Section
	45	1770-C5-22/3.2	Gate Tower - Base Unit - Horizontal Sections I
	46	1770-C5-22/27.3	Tunnel - Plan, Sections and Details
	47	1770-C5-10/3.1	Transition and Tunnel Support-Plan
	48	1770-C5-10/4	Tunnel and Transition Grouting Details
- -		1770-C5-22/28.4	Stilling Basin - Plans, Sections and Details
	50	1770-C5-45/1.1	Service Bridge - Plan and Elevation
	51	·	Gate Tower - Stability Analysis
	52		Tunnel Section Design Analysis
	53	1770-DM9-98/13	Materials Usage Chart

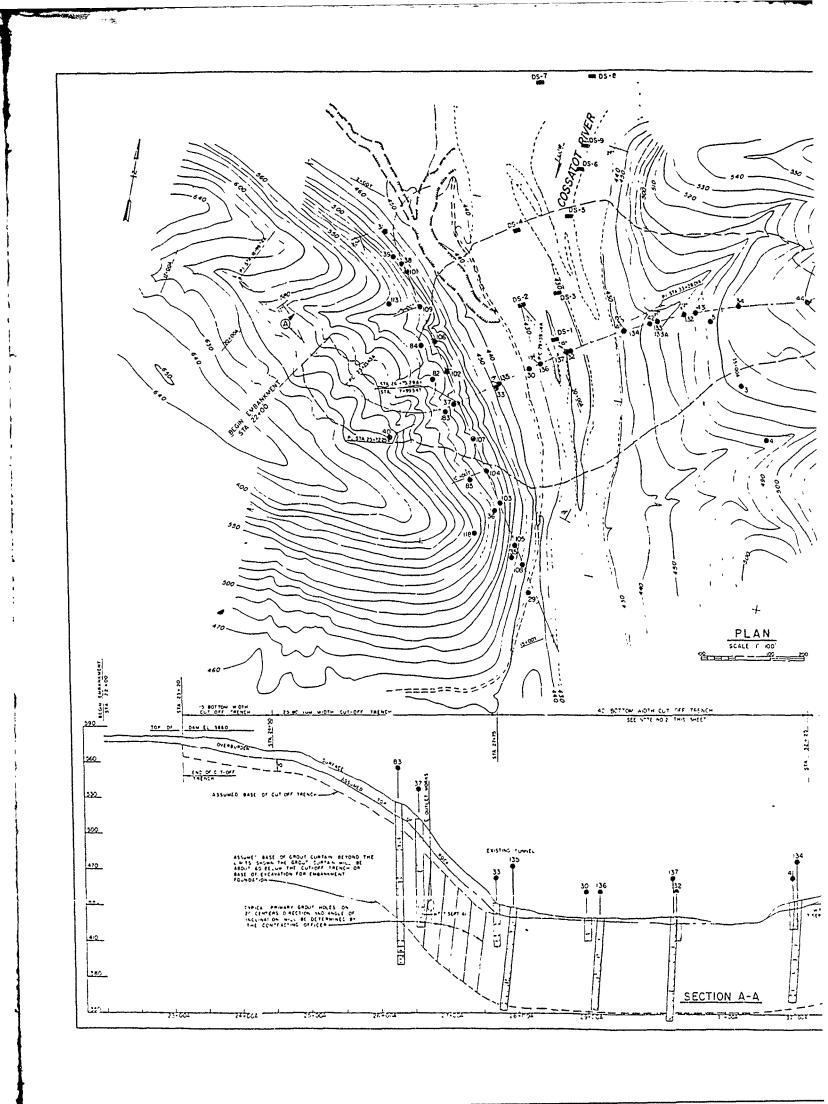


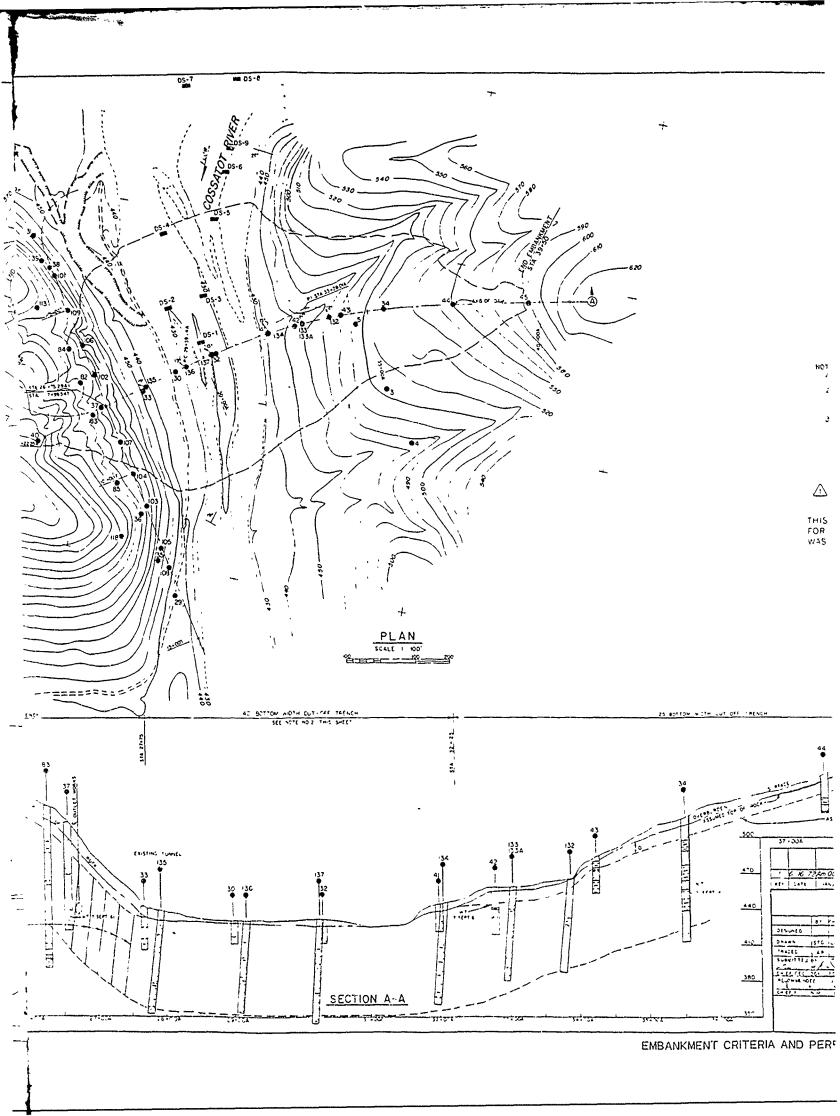


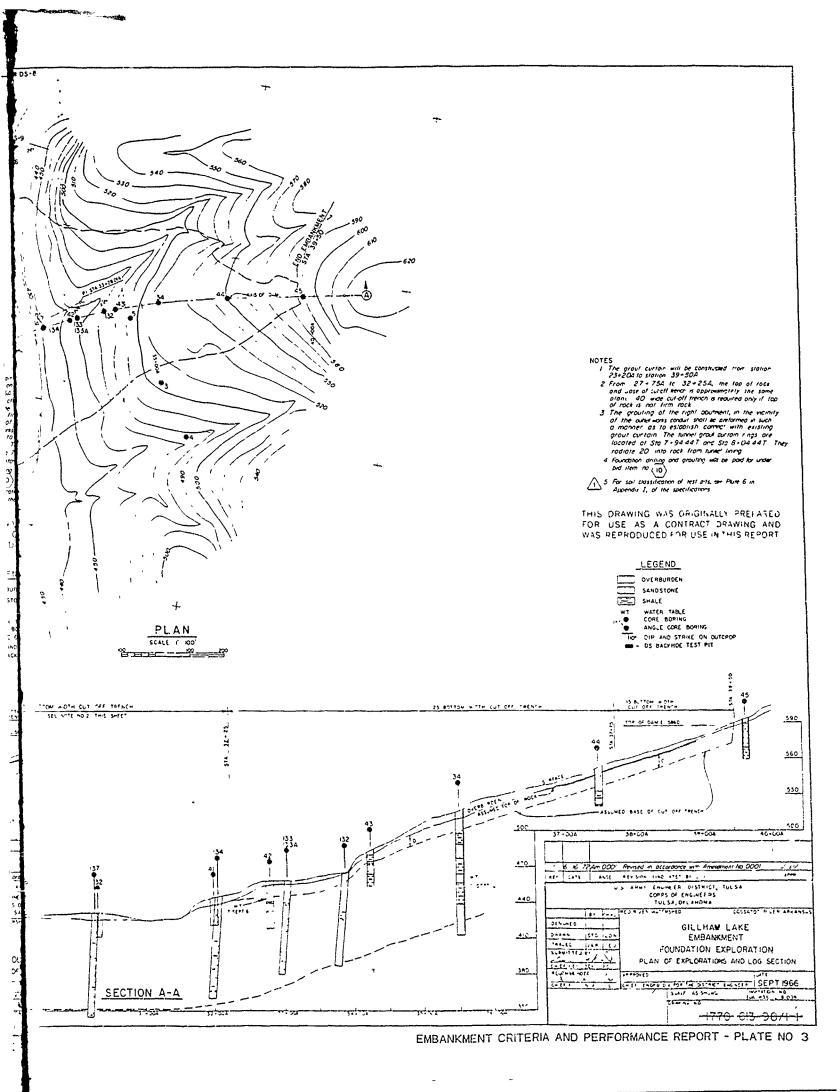


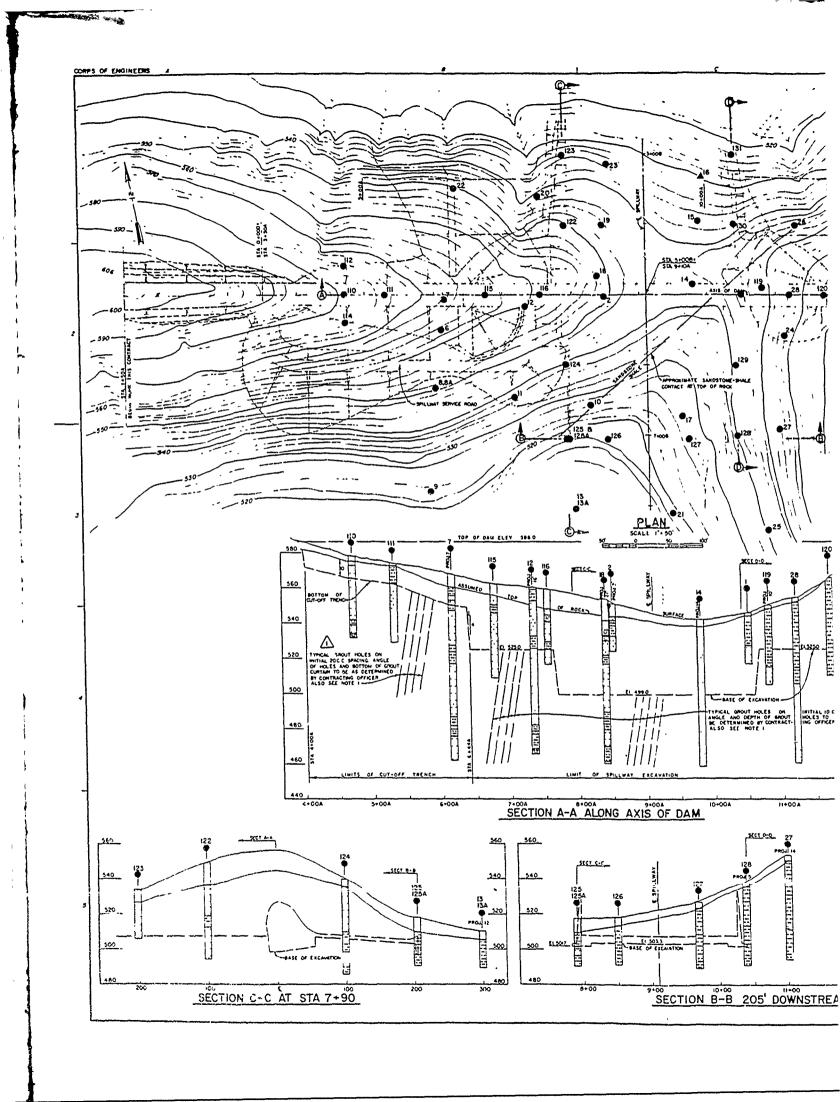


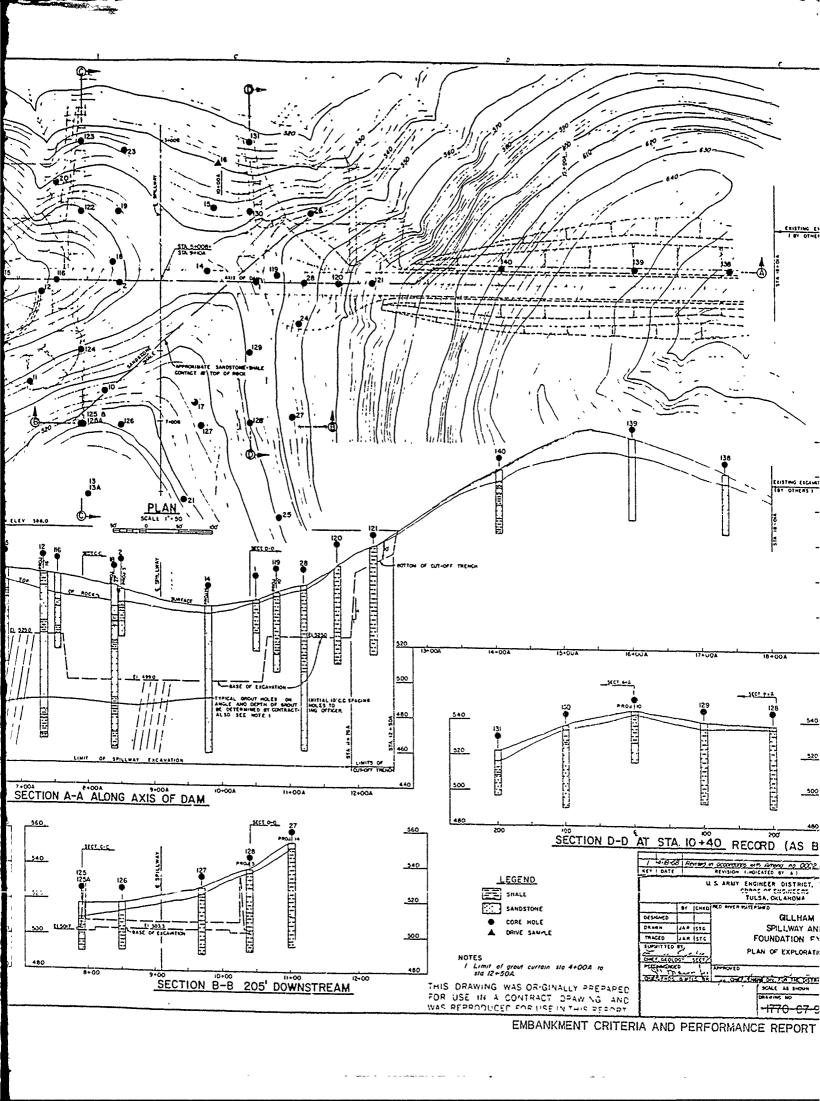


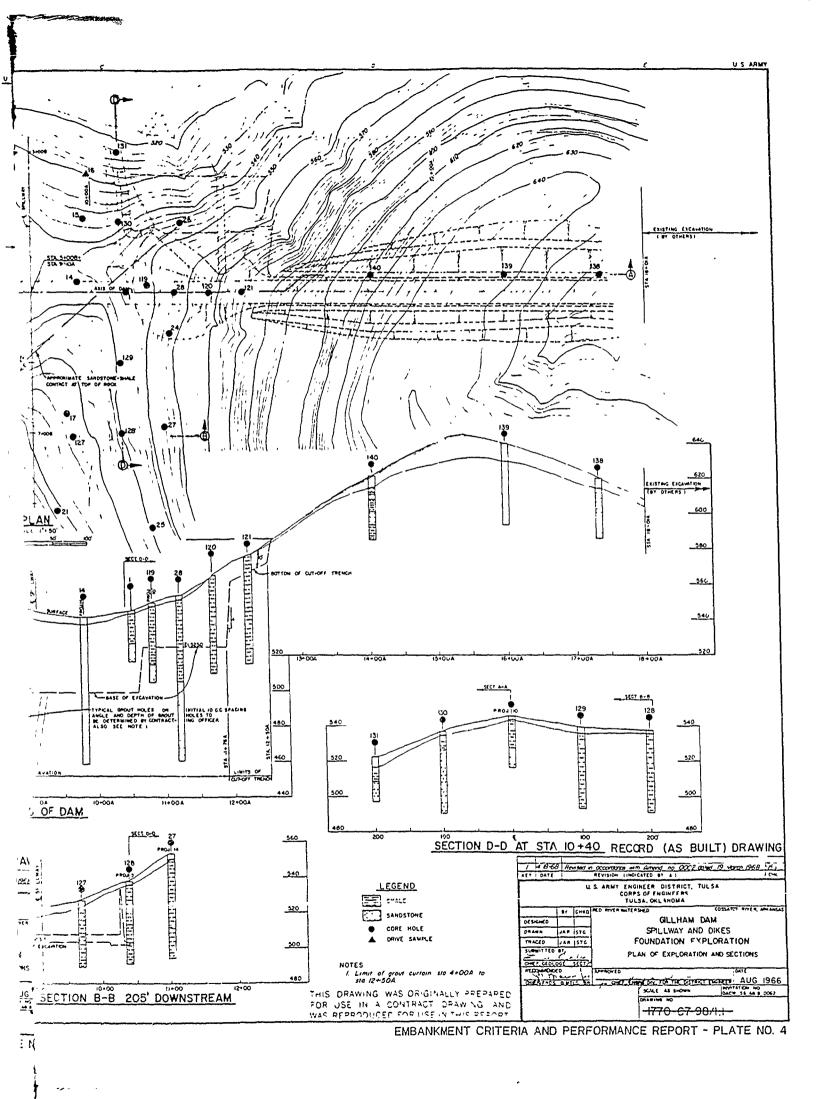


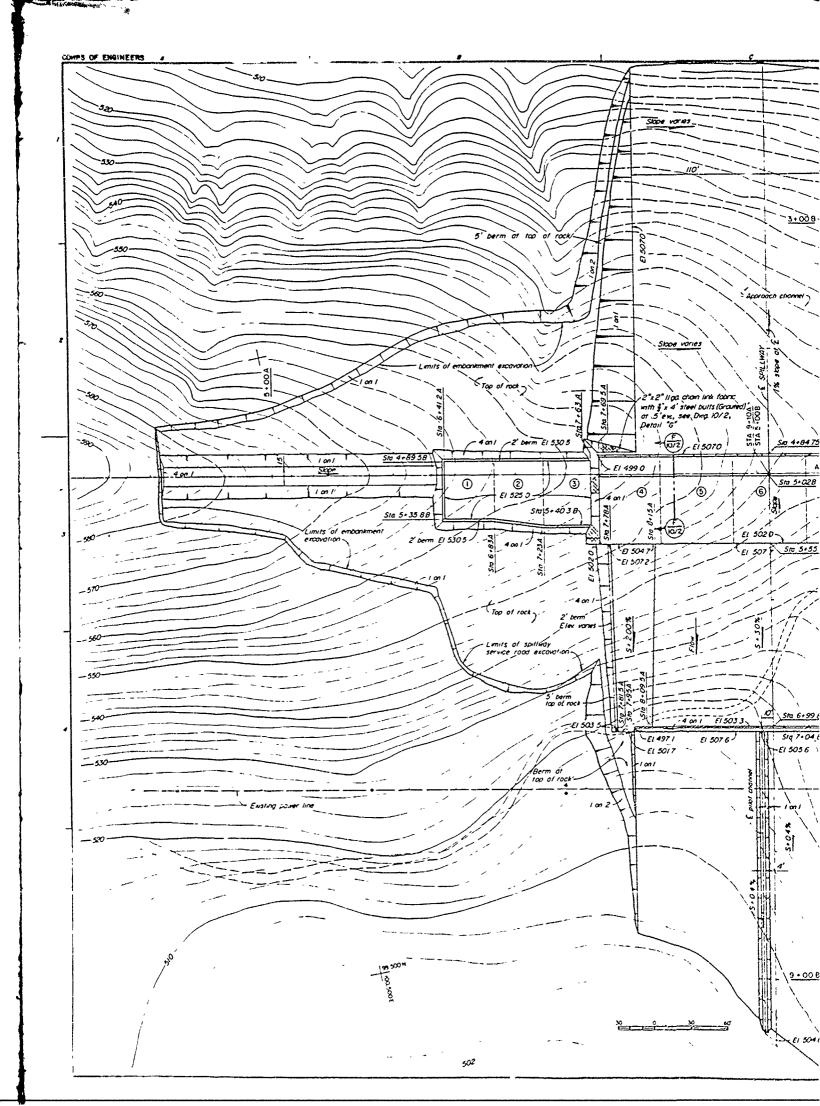


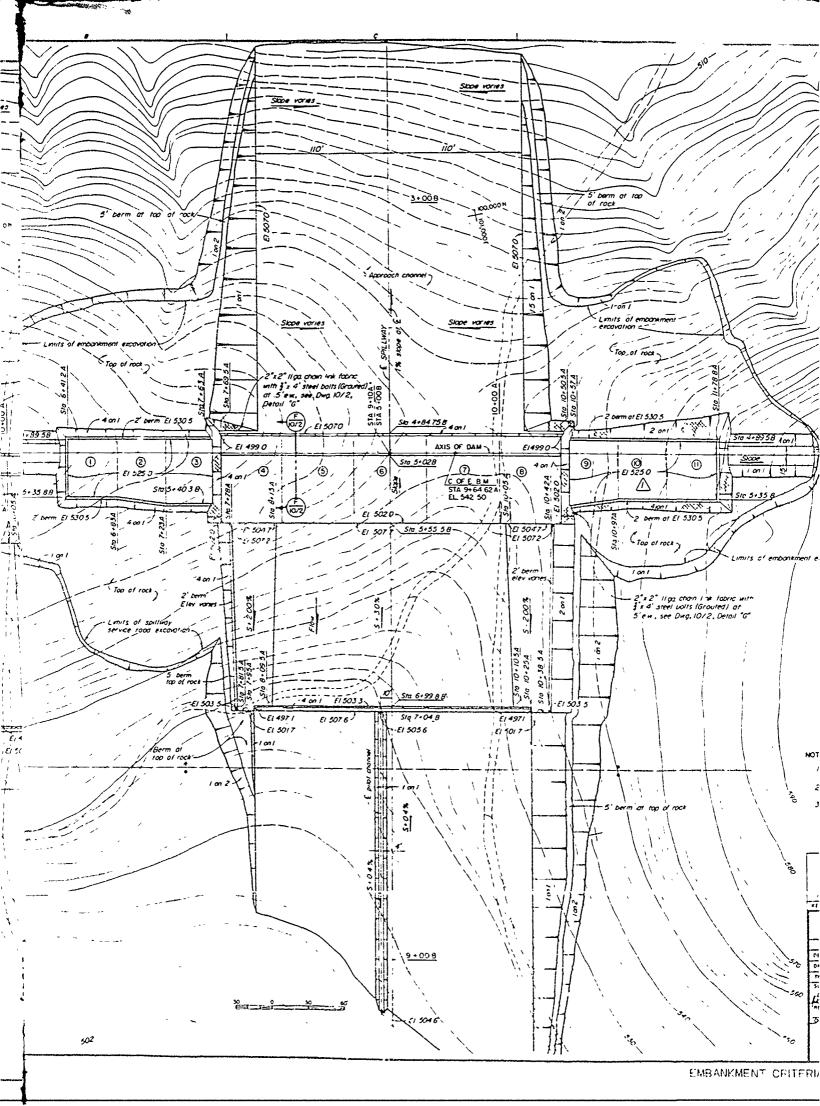


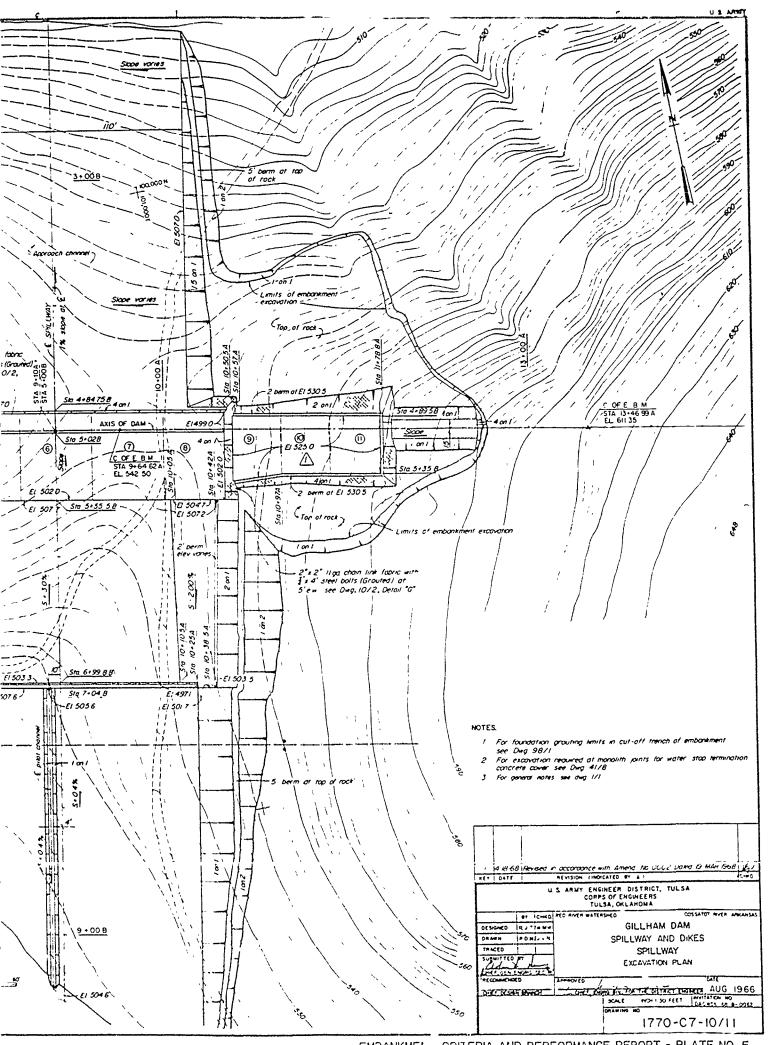










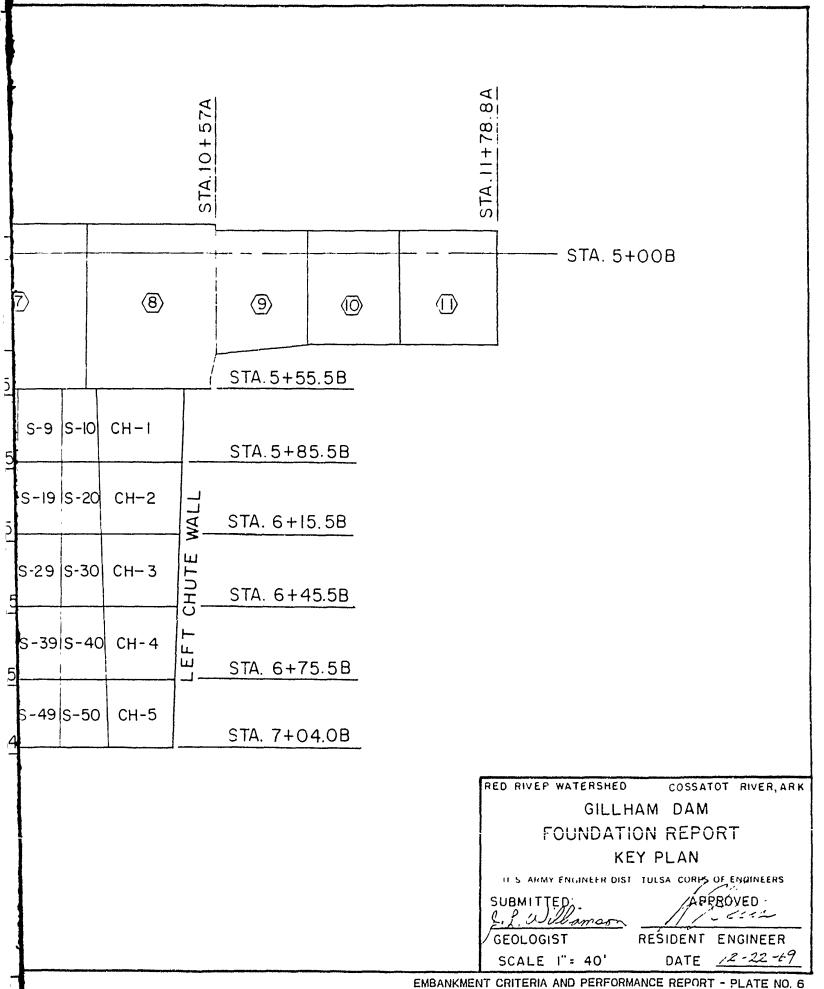


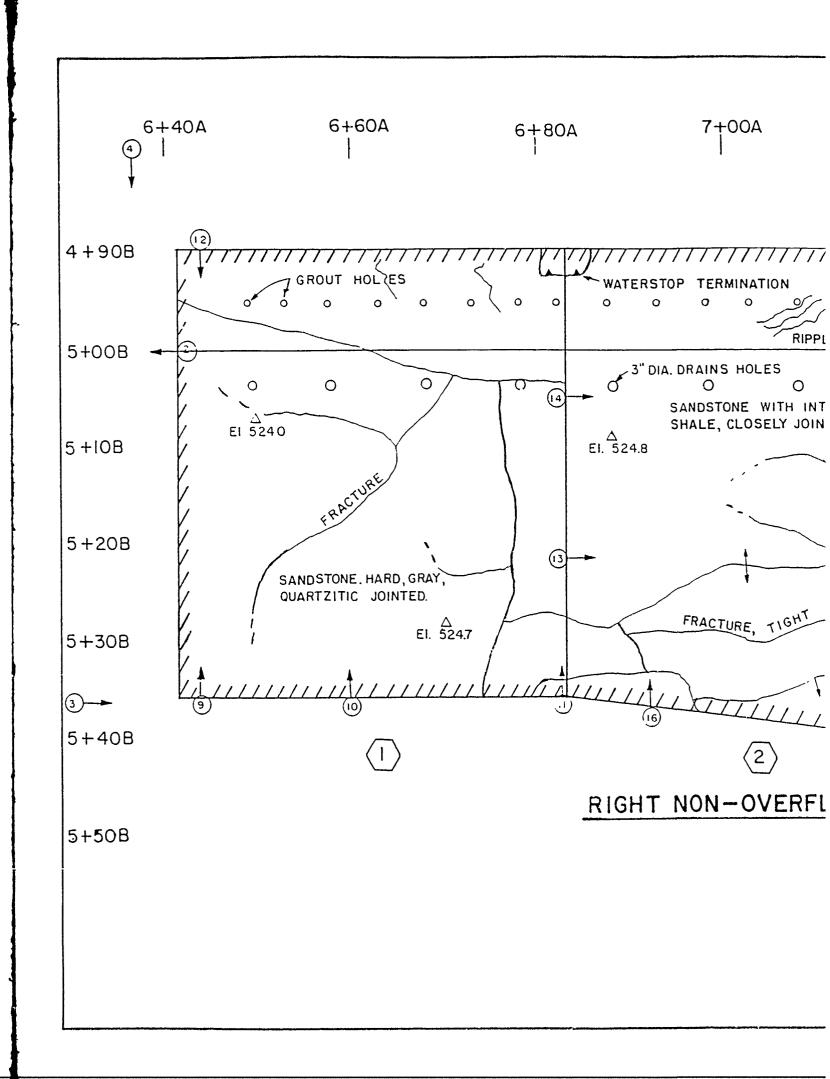
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	y)			CHUTE	CH-3	S-2	I S-22	S-23	S-24	S-25	S-26
		ſ		RIGHT (CH-4	s-3	I S-32	S-33	S-34	S-35	S-36
				L	CH-5	s-3	I S-42	S-43	s . 44	S-45	S-46
									-		

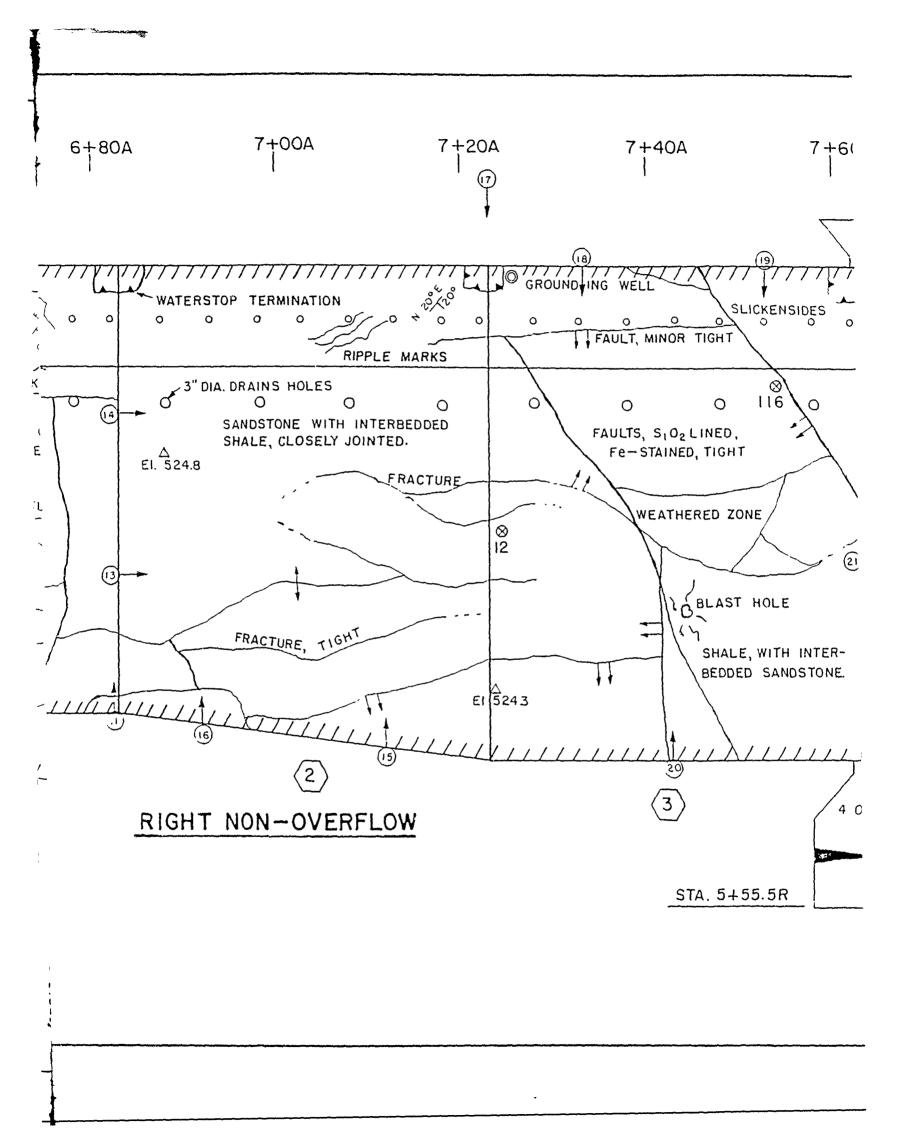
The state of the s

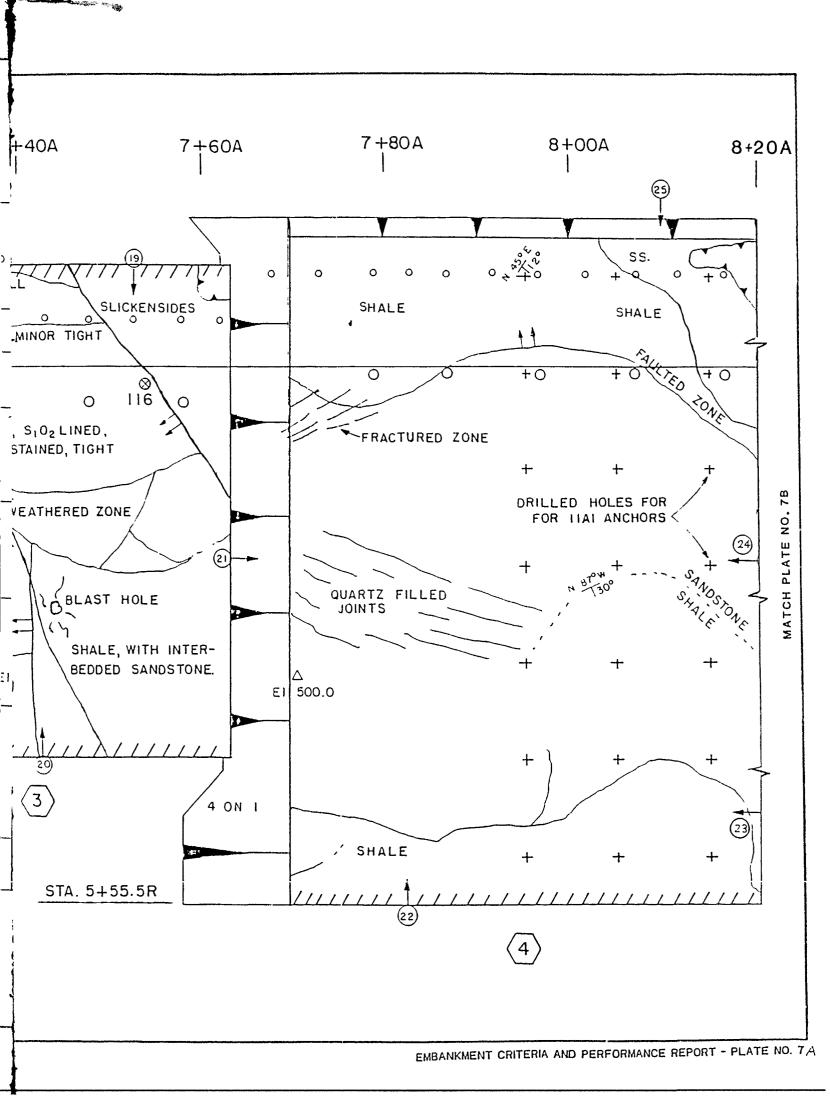
				-		-										
		£ SPILLWAY														
	STA. 7+63A							9+10A							STA.10+57A	į
	STA							STA.		<u></u>	-				STA.	!
							A	KIS O	F DA	M						
	3	4			(5)			6			7			8		1
						V						·			/	ST
-			CH-1	S-	I S-2	S-3	S-4	S-5	S-6	S-7	S-8	S-9	S-10	CH-1		ST
•		WALL	CH-2	S-I	S-12	S-13	S-I4	S-I5	S-16	S-I7	S-18	S-19	S-20	CH-2	WALL	ST
,		CHUTE \	CH-3	S-2	.I S-22	S-23	S-24	S-25	S-26	S-27	S-28	S-29	s-30	CH-3	CHUTE V	ST
7		RIGHT C	CH-4	s-3	S-32	S-33	S-34	S-35	S-36	S-37	S-38	S-39	S-40	CH-4	LEFT C	ST
		u	CH-5	S-3	I S-42	S-43	S-44	S-45	S-46	S-47	S-48	S-49	S~50	CH-5		ST/
,			\	*		<u> </u>		······································		<u> </u>	<u> </u>				• —	

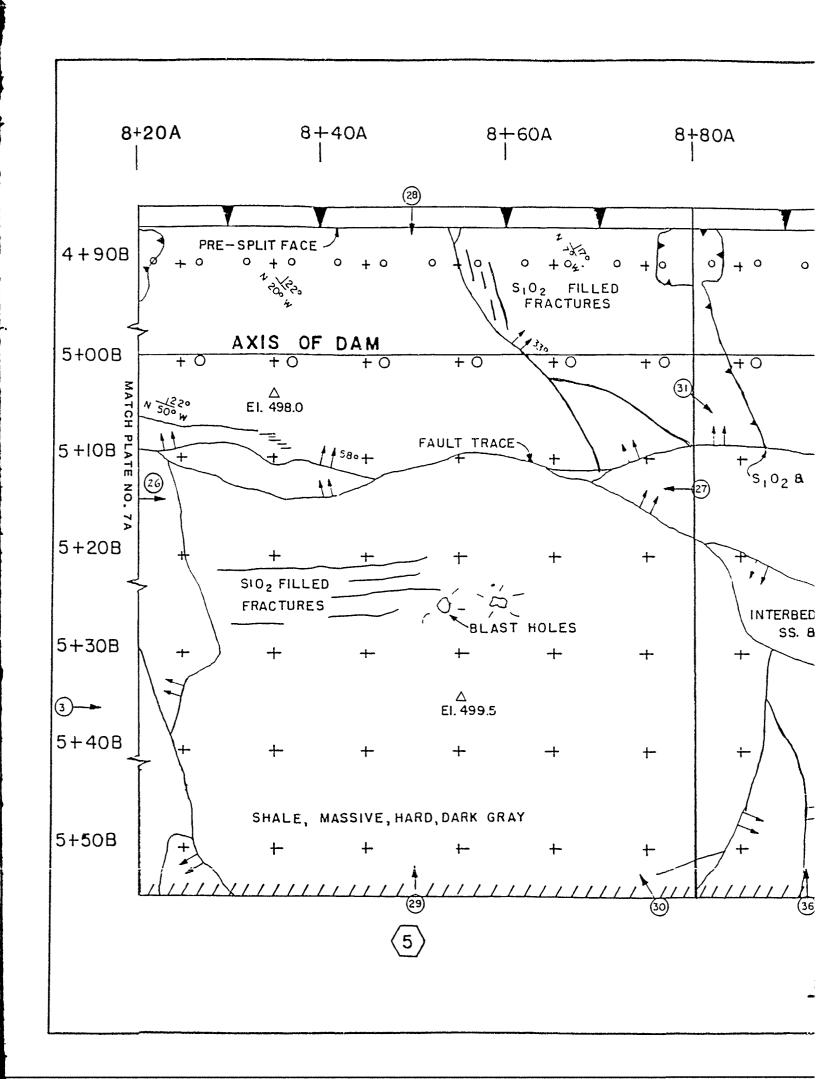
KEY PLAN

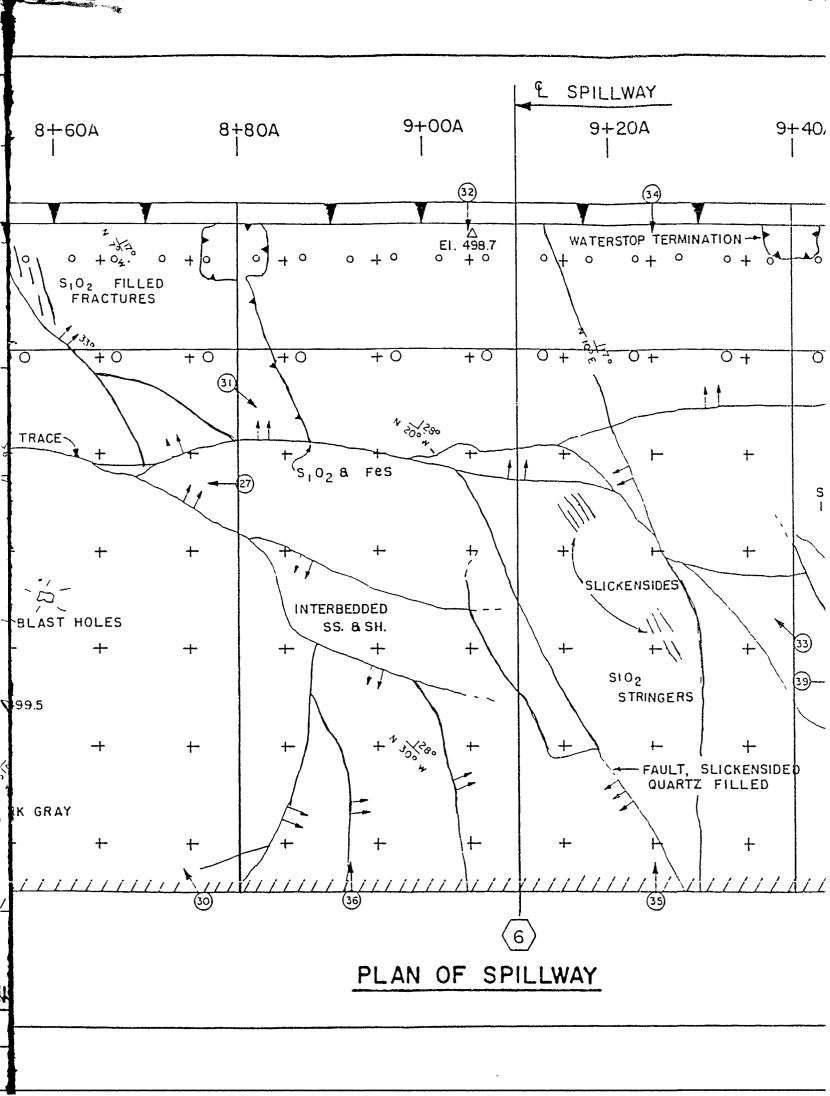


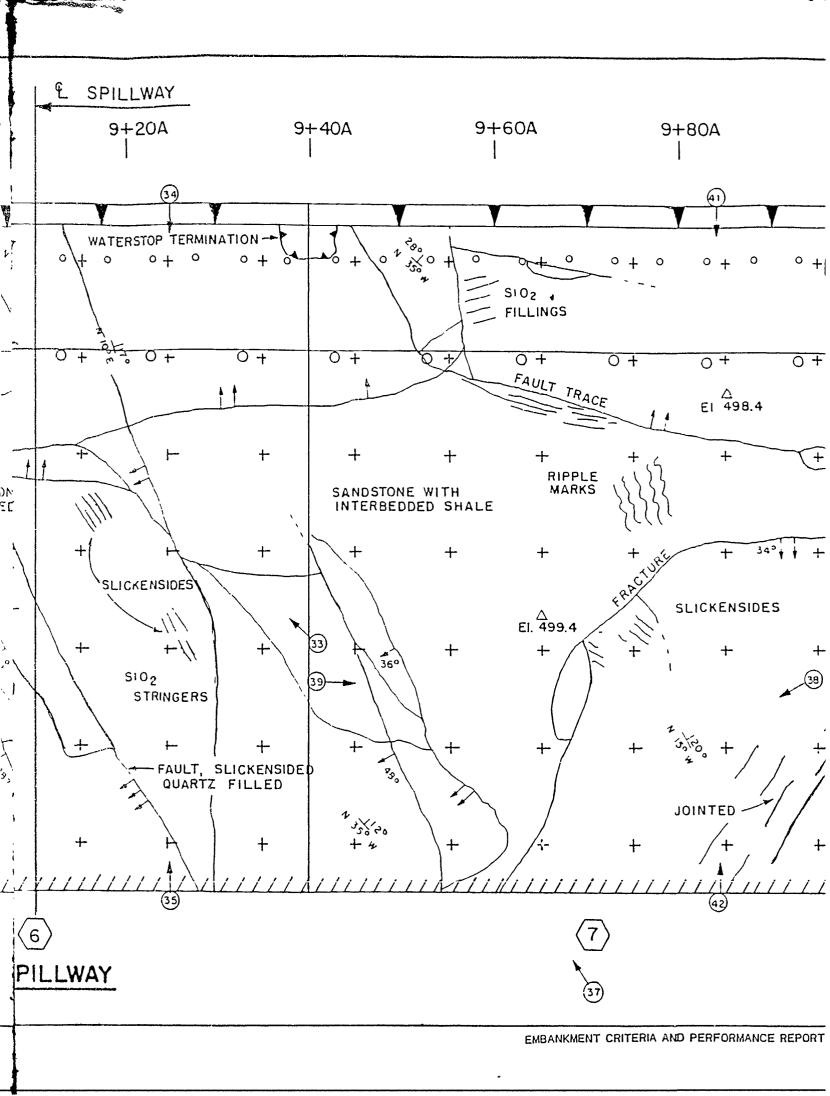


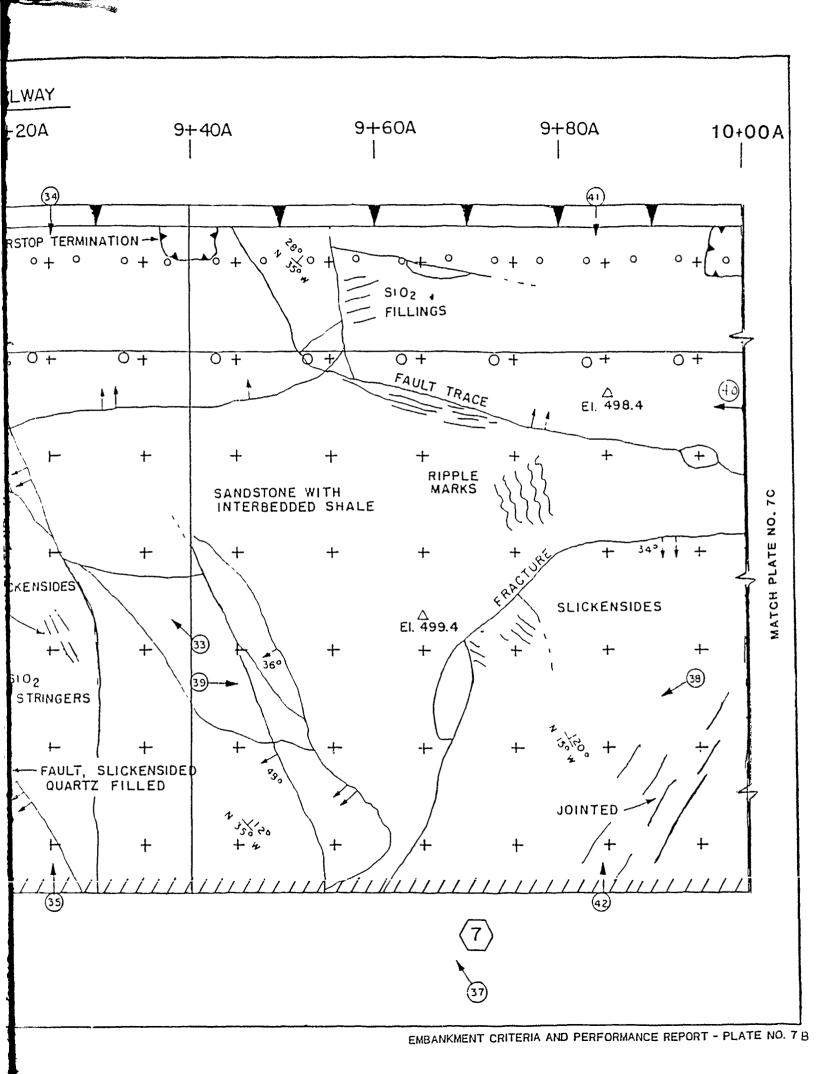


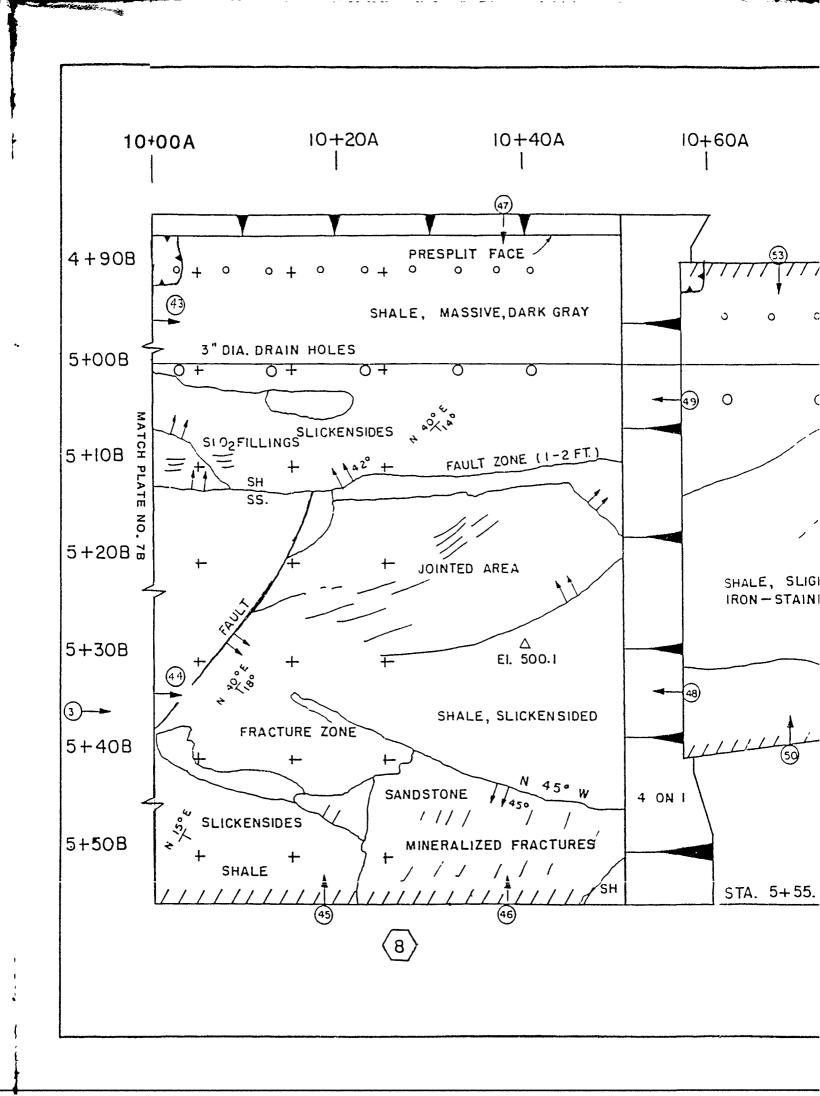


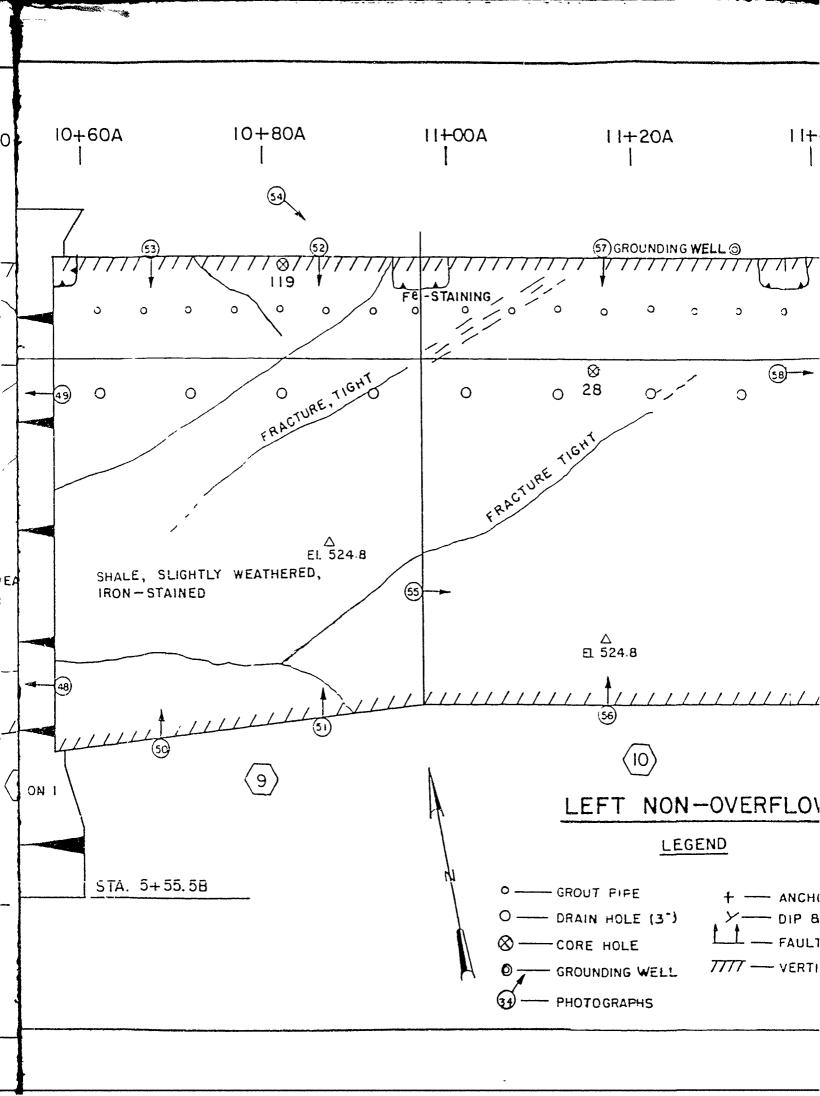


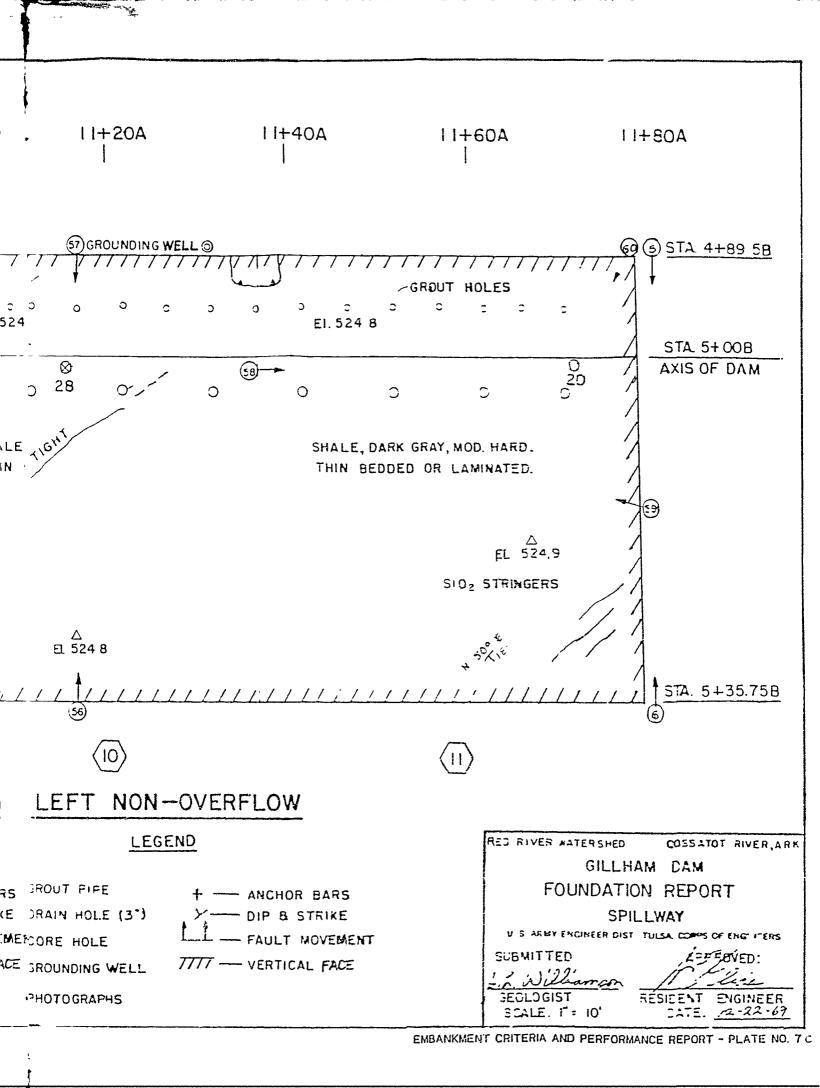


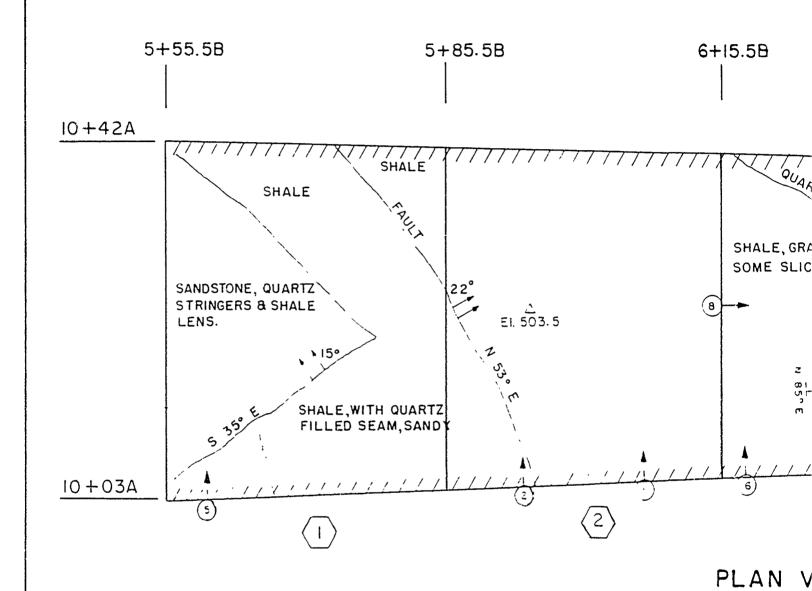




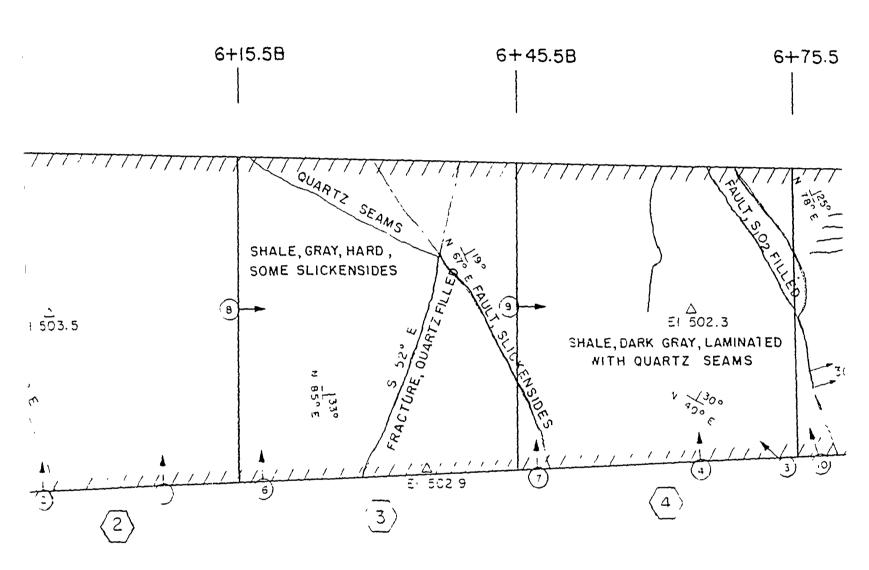




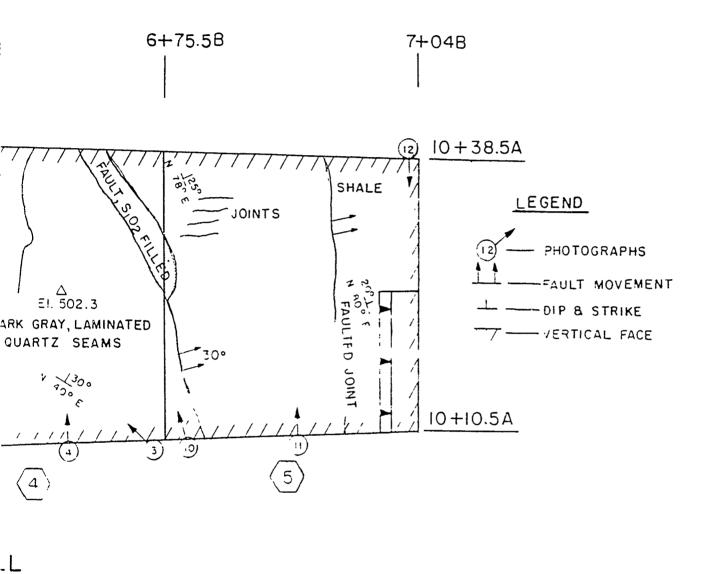








PLAN VIEW-' EFT CHUTE WALL



GILLHAM DAM

FOUNDATION REPORT

LEFT CHUTE WALL

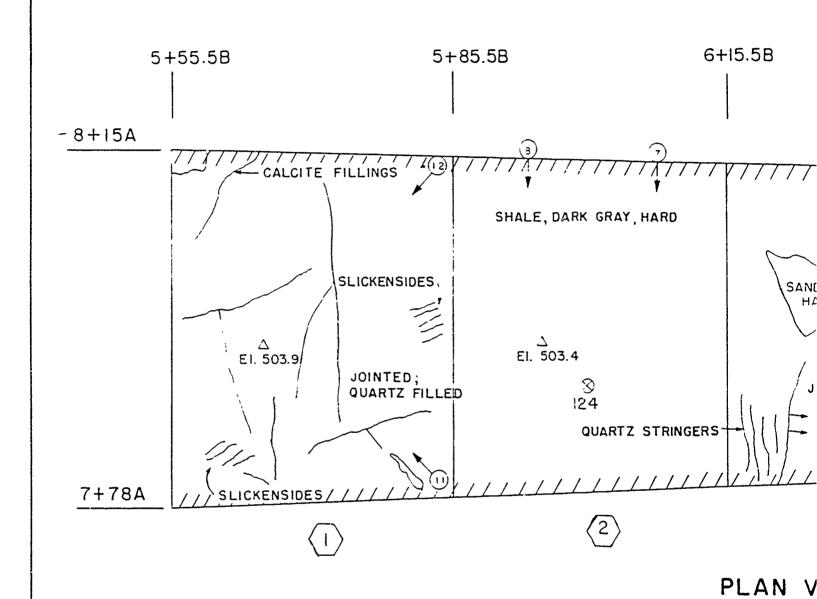
II S ARMY ENGINEER DIST TULSA CORPS OF ENGINEERS

SUBMITTED:

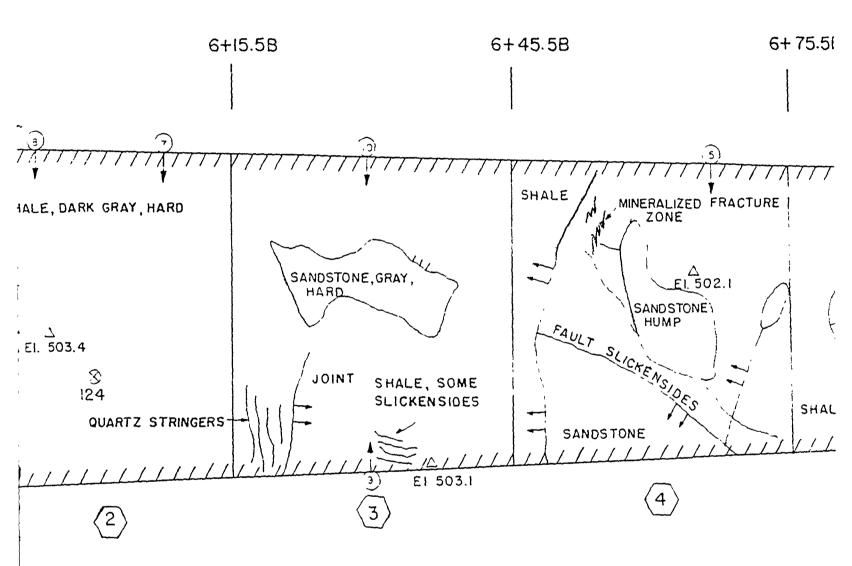
JEOLOGIST

SCALE I" = 10'

DATE: 12-22-69

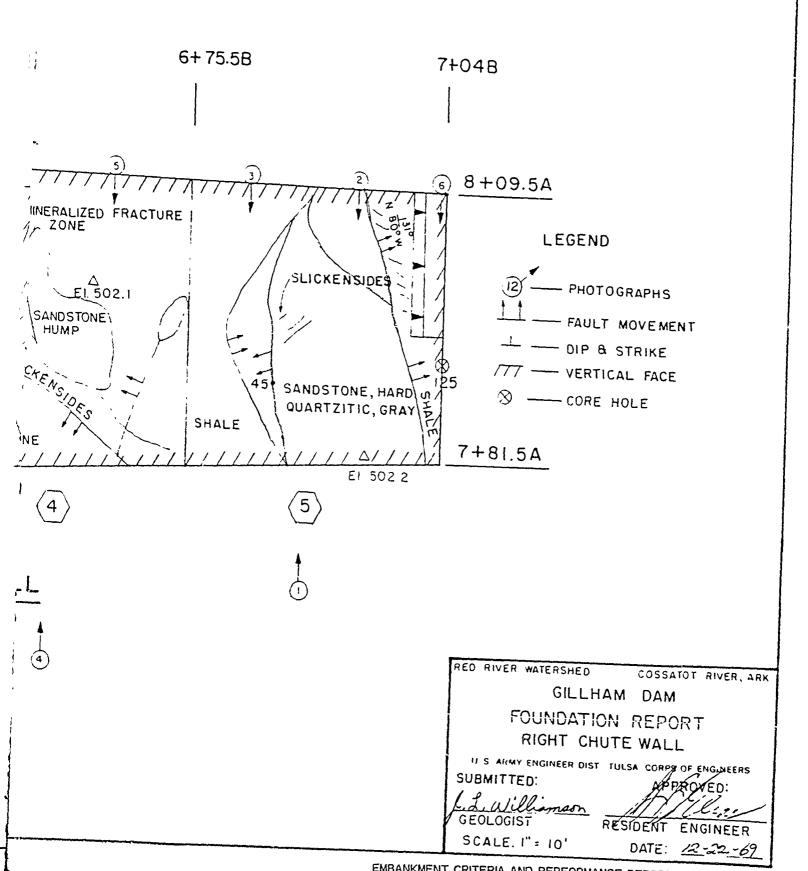


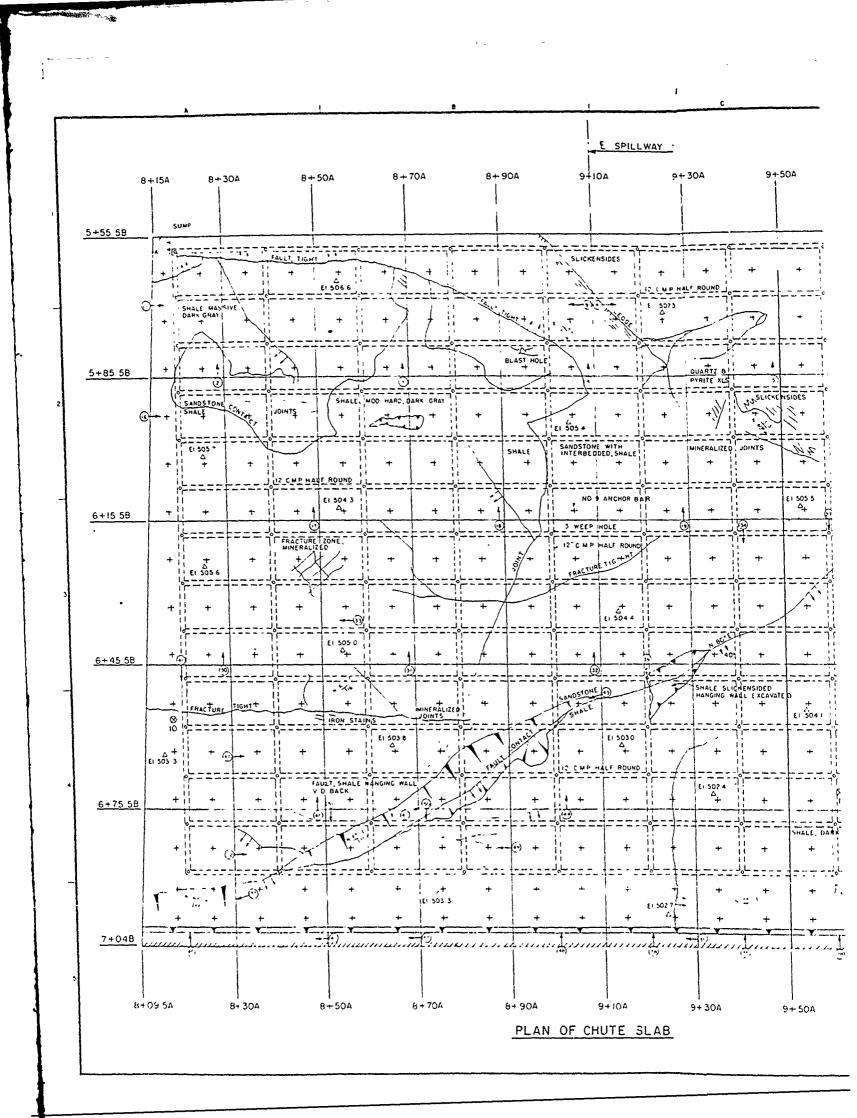


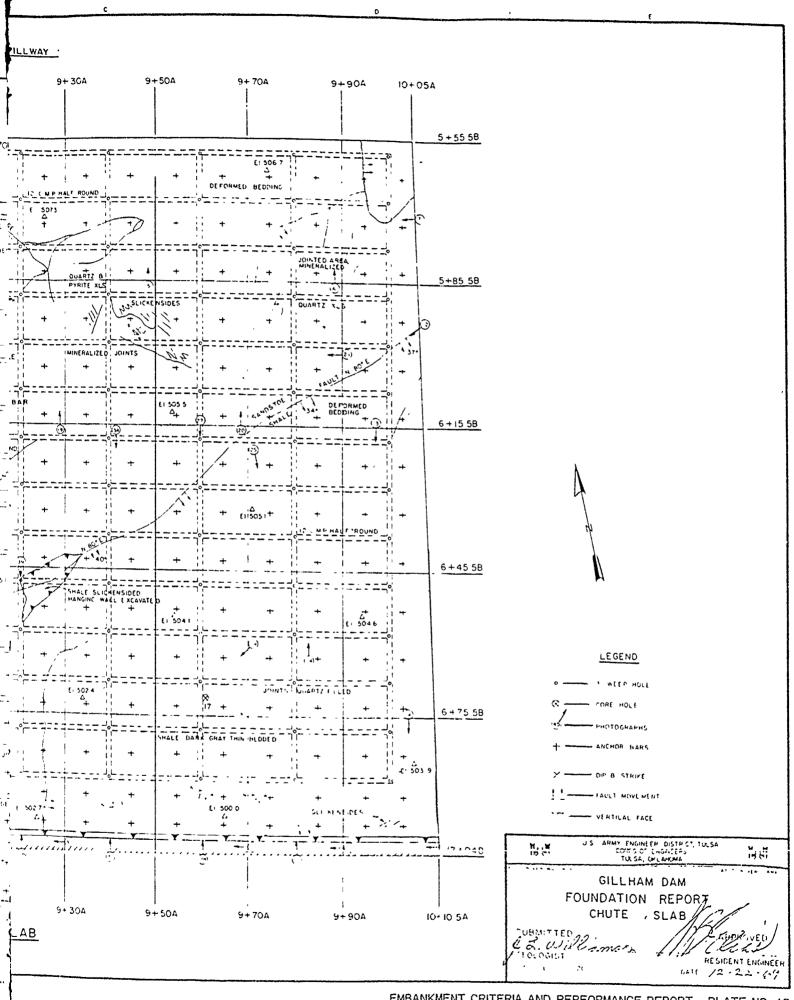


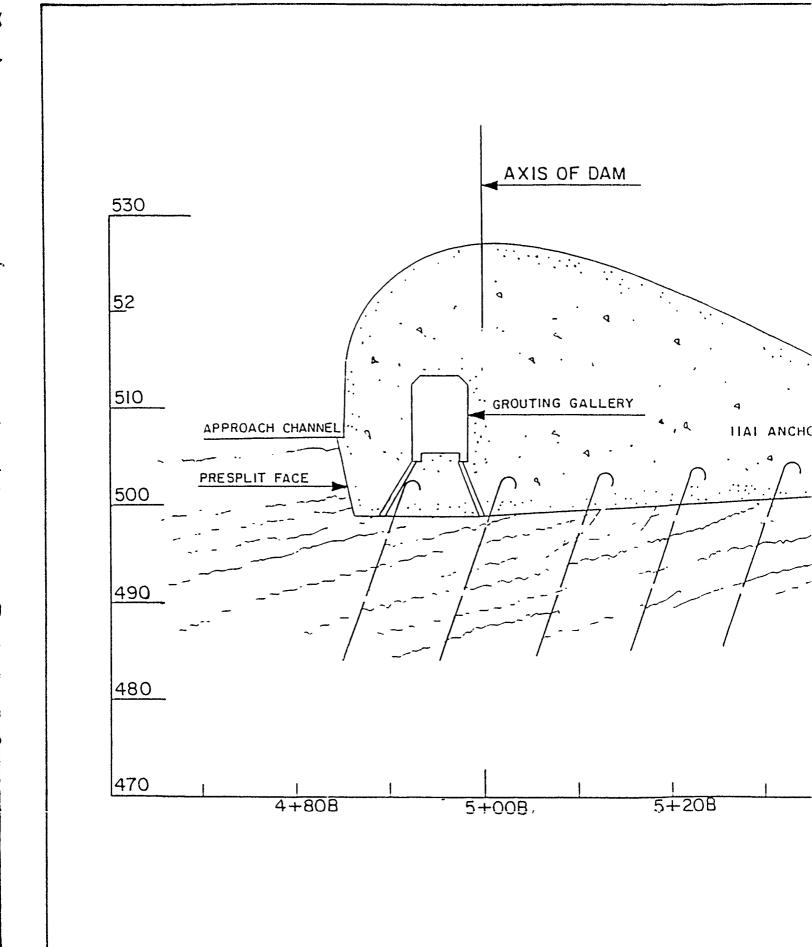
PLAN VIEW-RIGHT CHUTE WALL

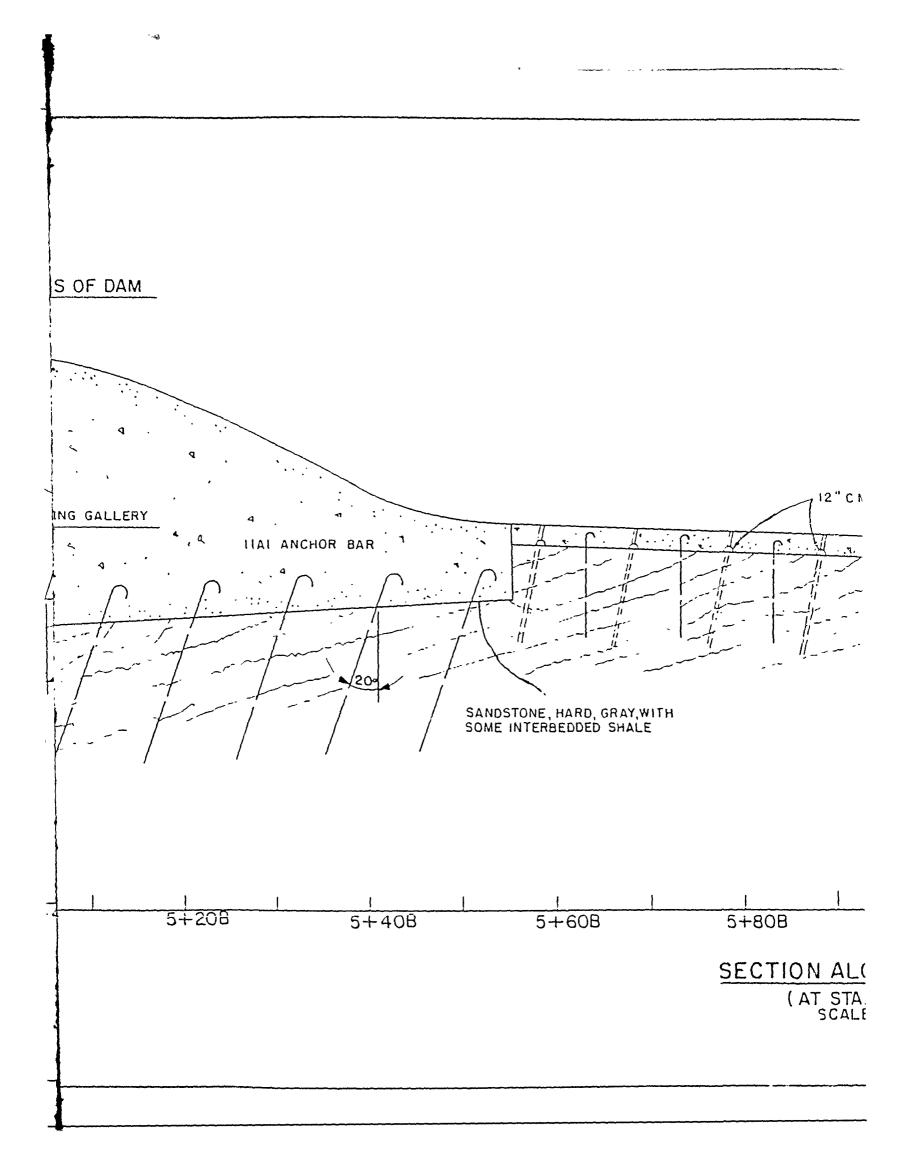
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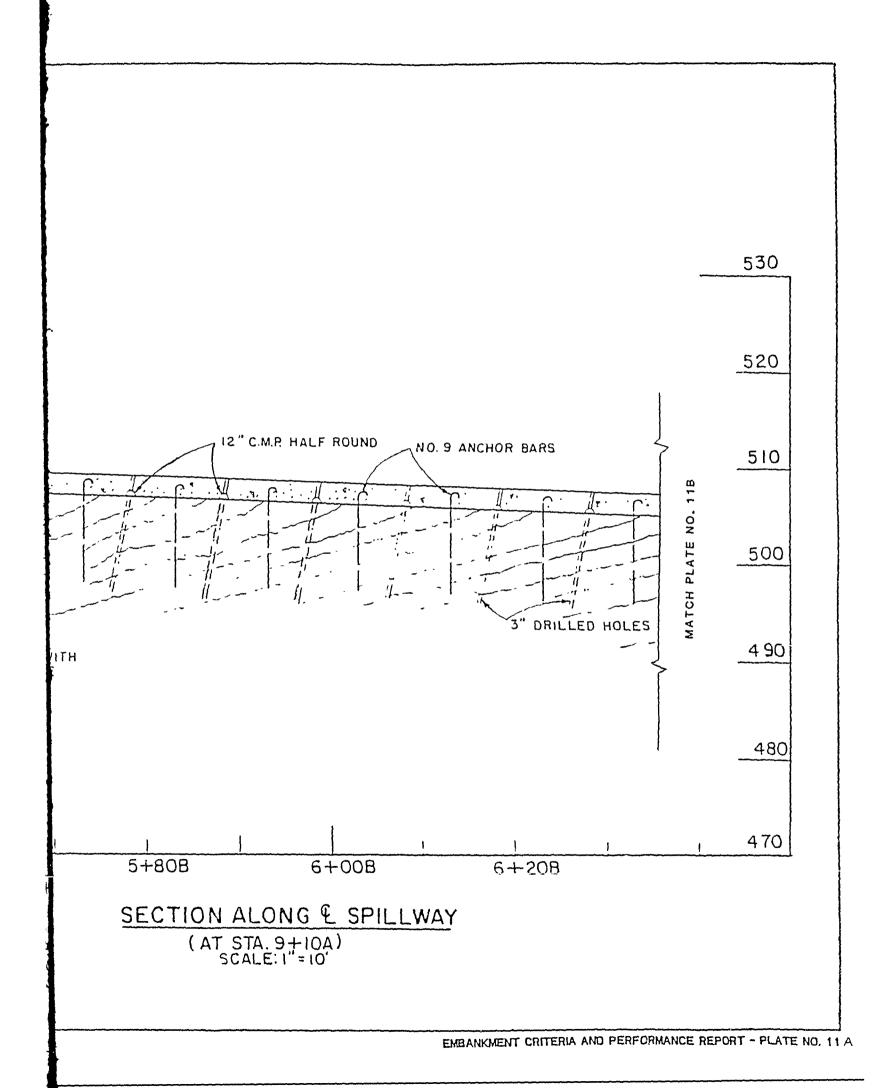


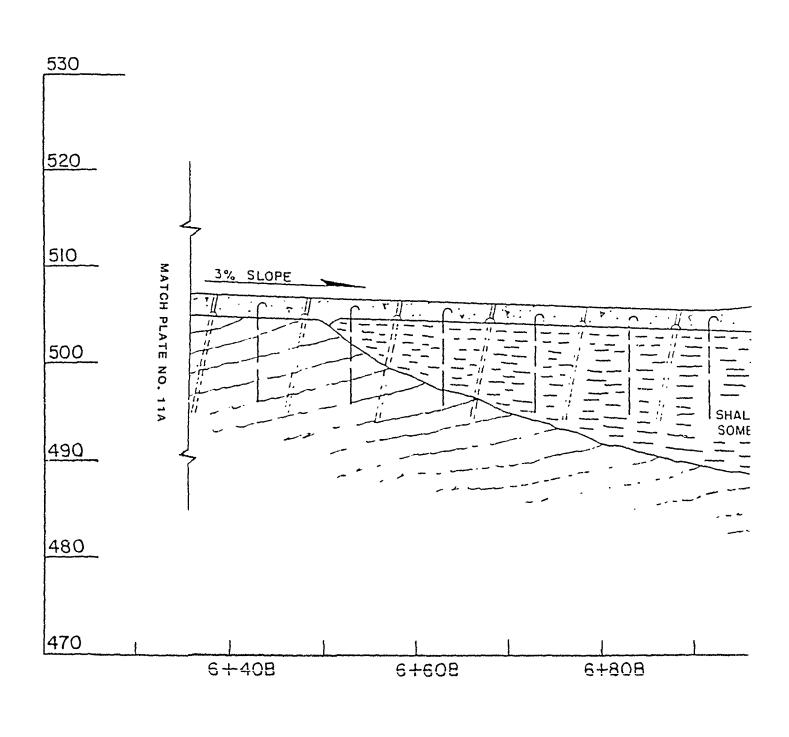


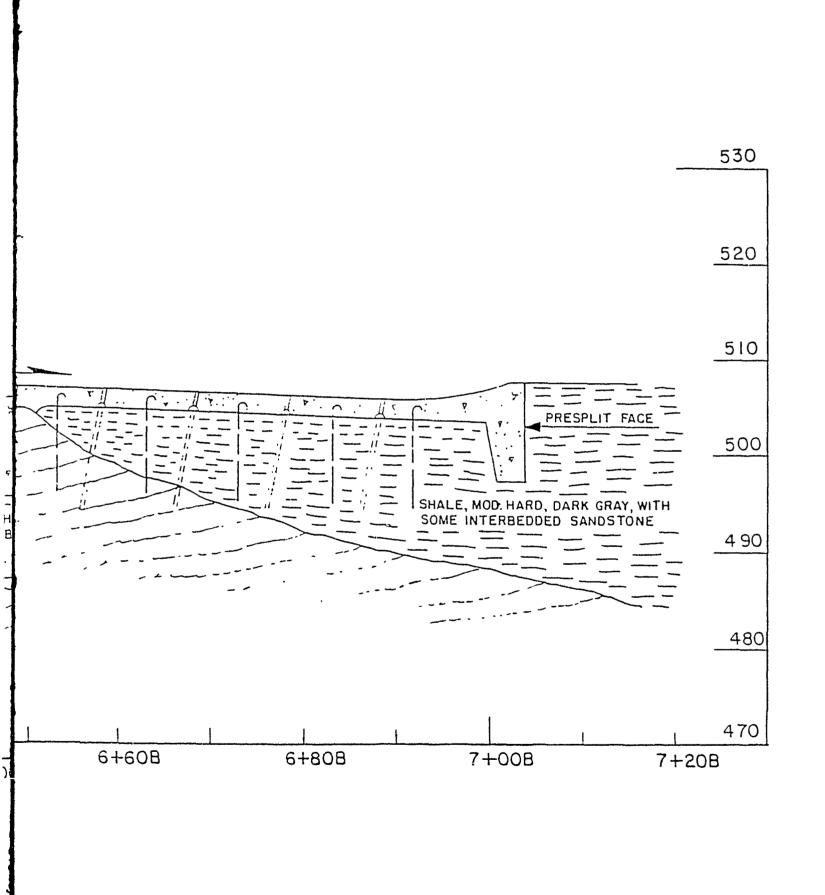


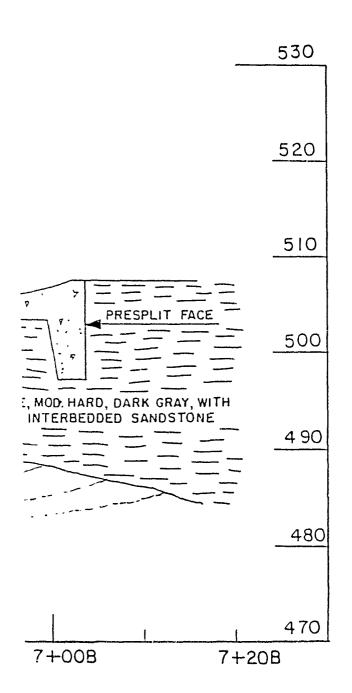




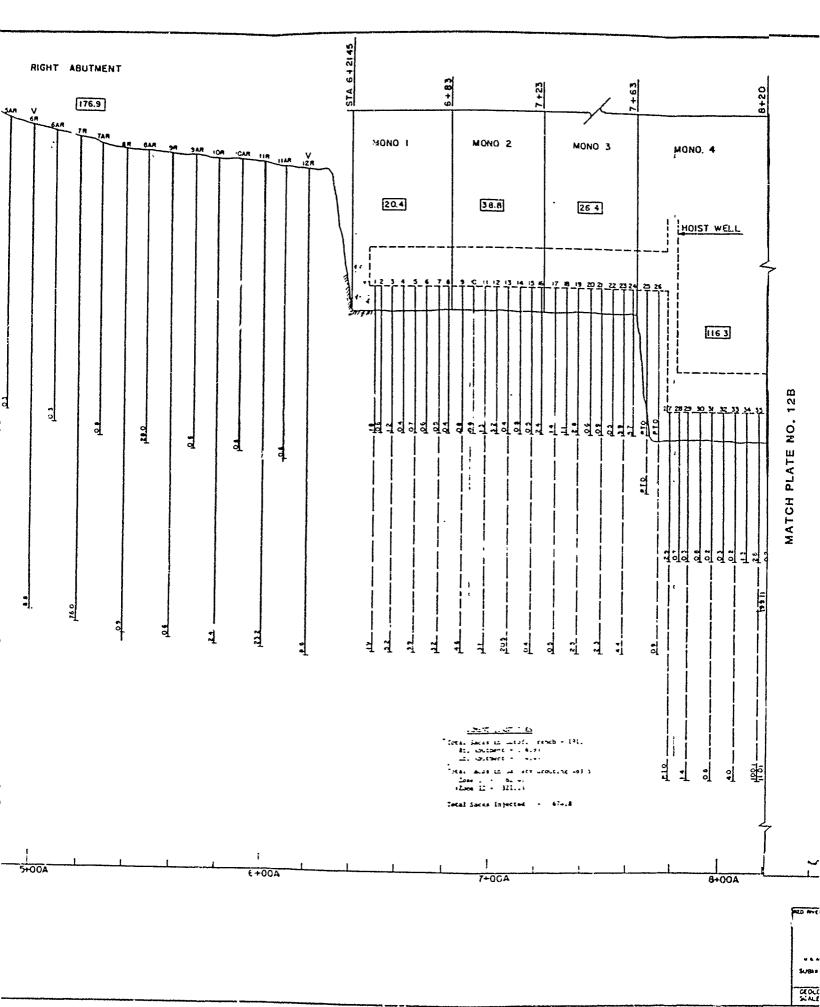


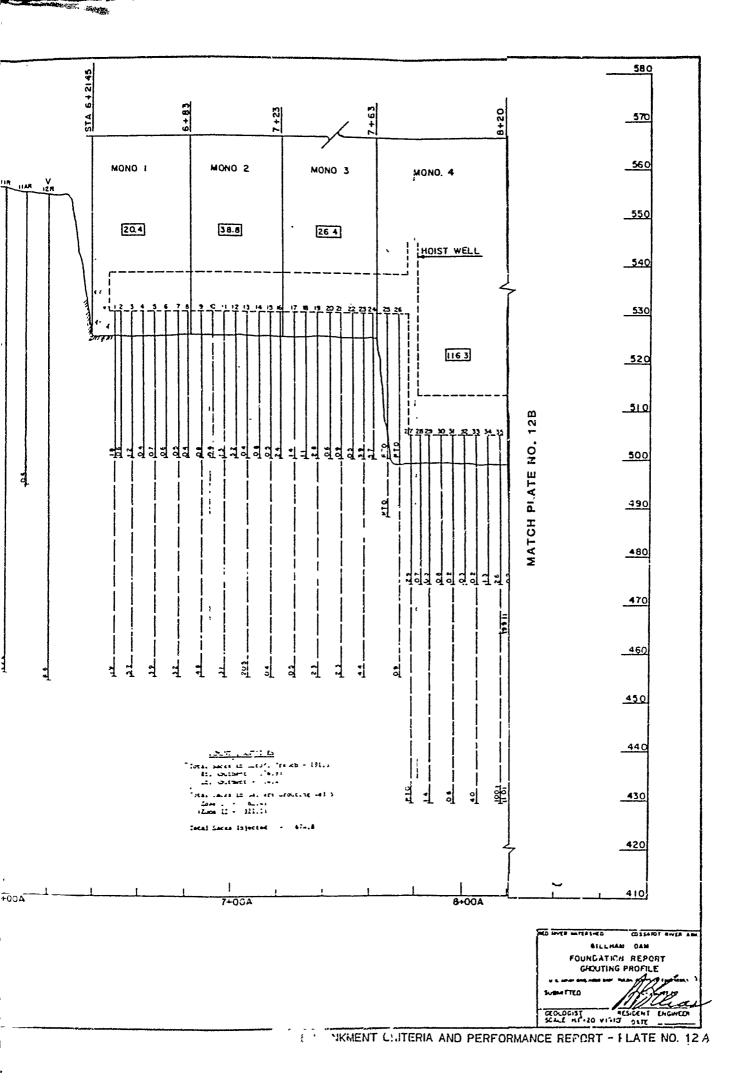


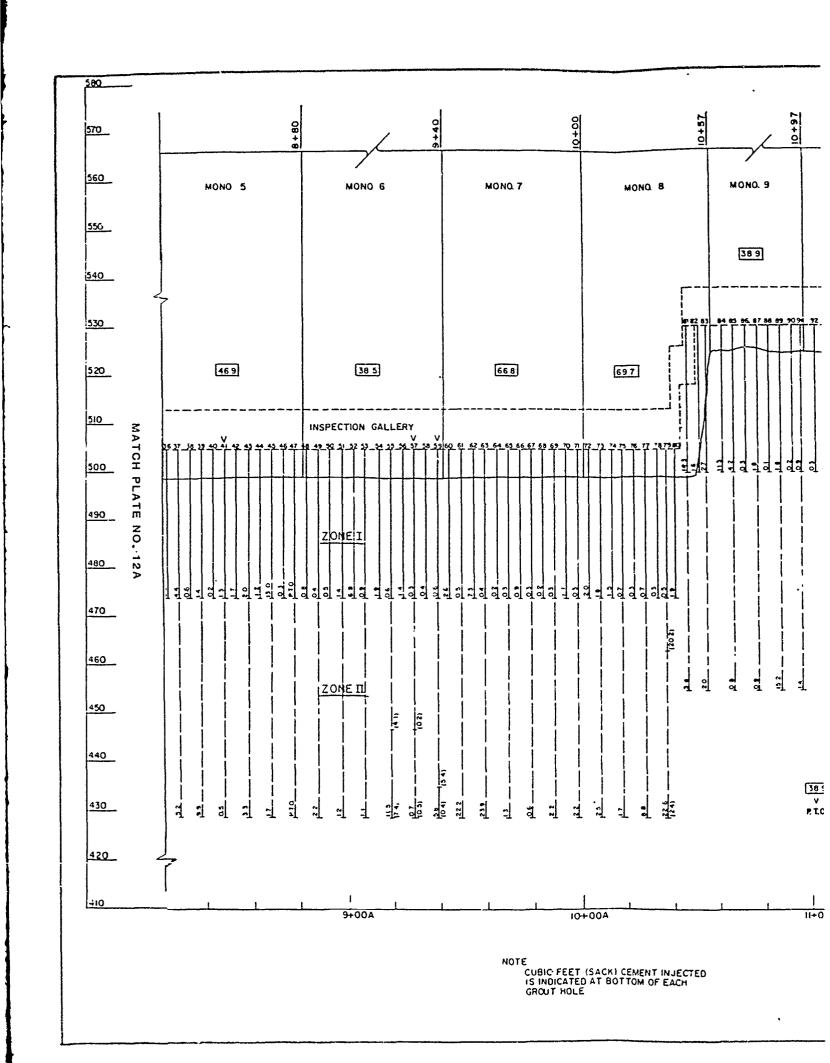


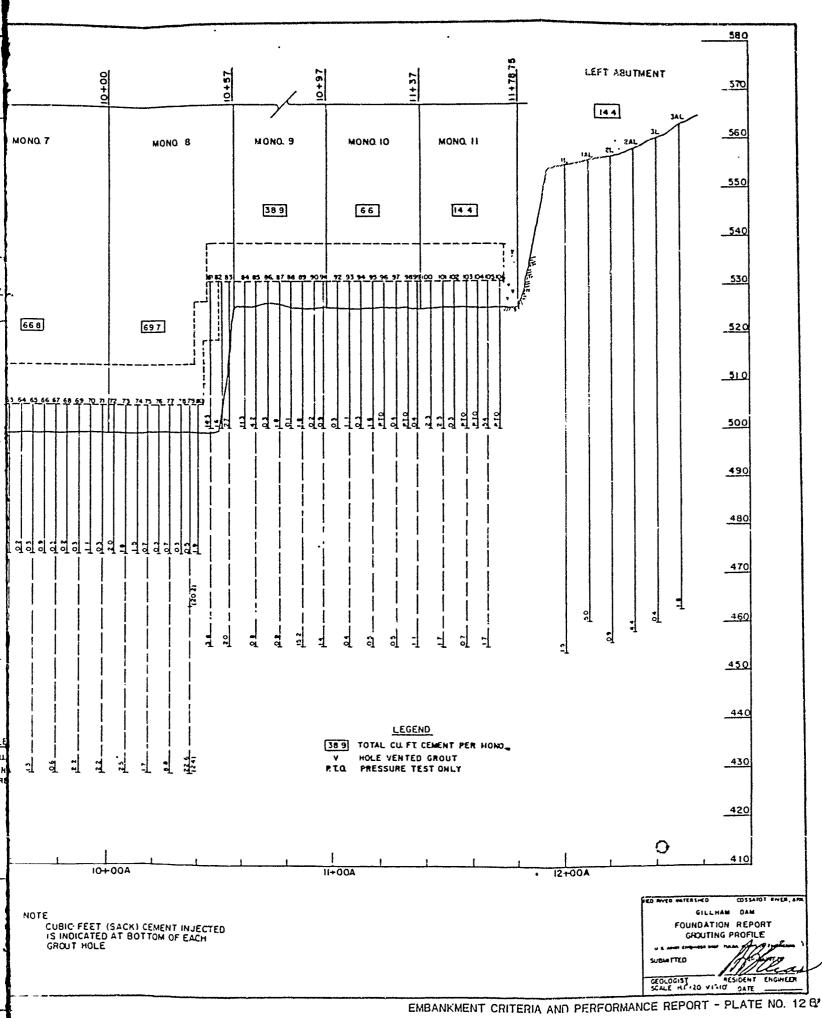


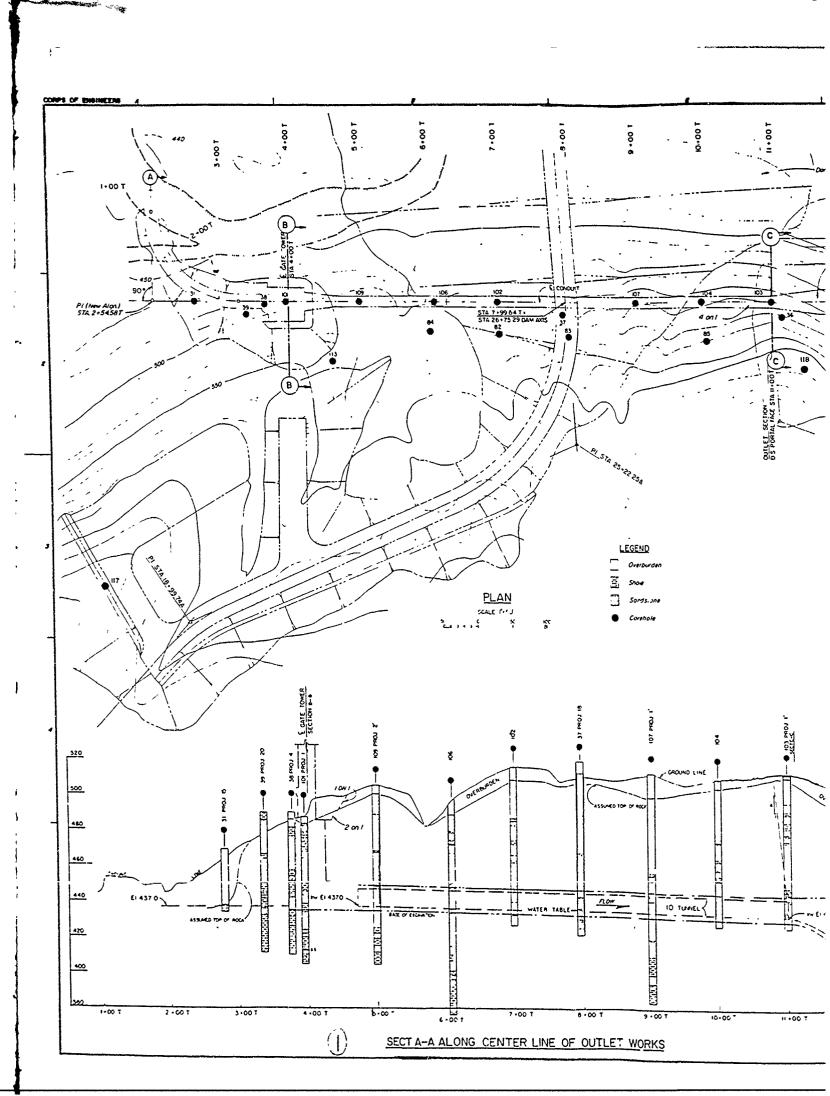
GILLHAM DAM
FOUNDATION REPORT
SECTION SPILLWAY & CHUTE SLAB
US ARMY ENGINEER DIST TULSA. CORPS OF ENGINEERS
SUBMITTED:
APPROVED:
APPROVED:
GEOLOGIST RESIDENT ENGINEER
SCALE. I" = 10' DATE: 12-22-69

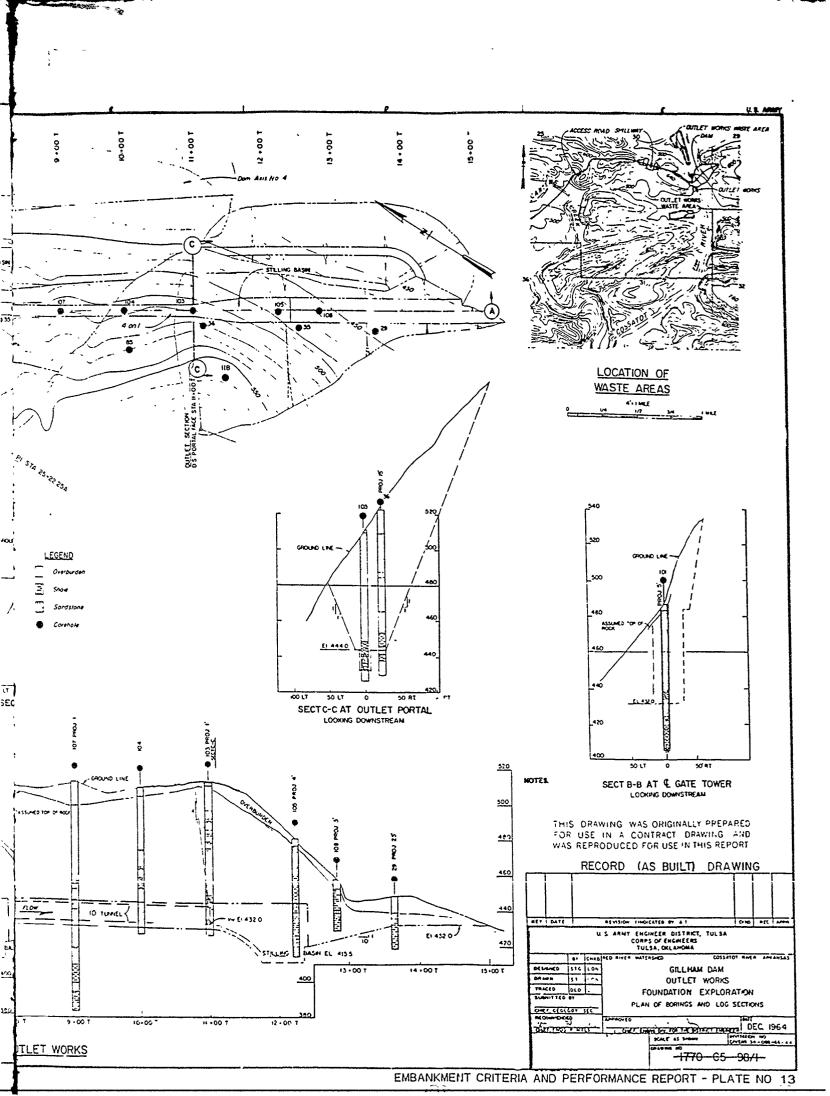


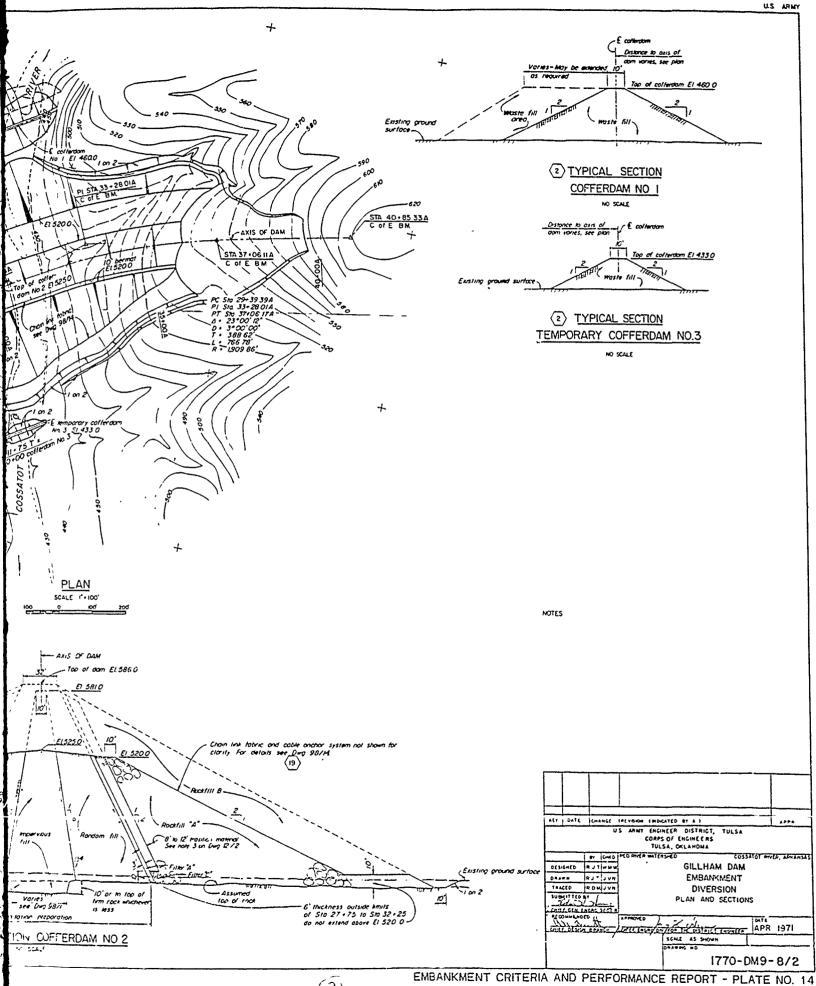


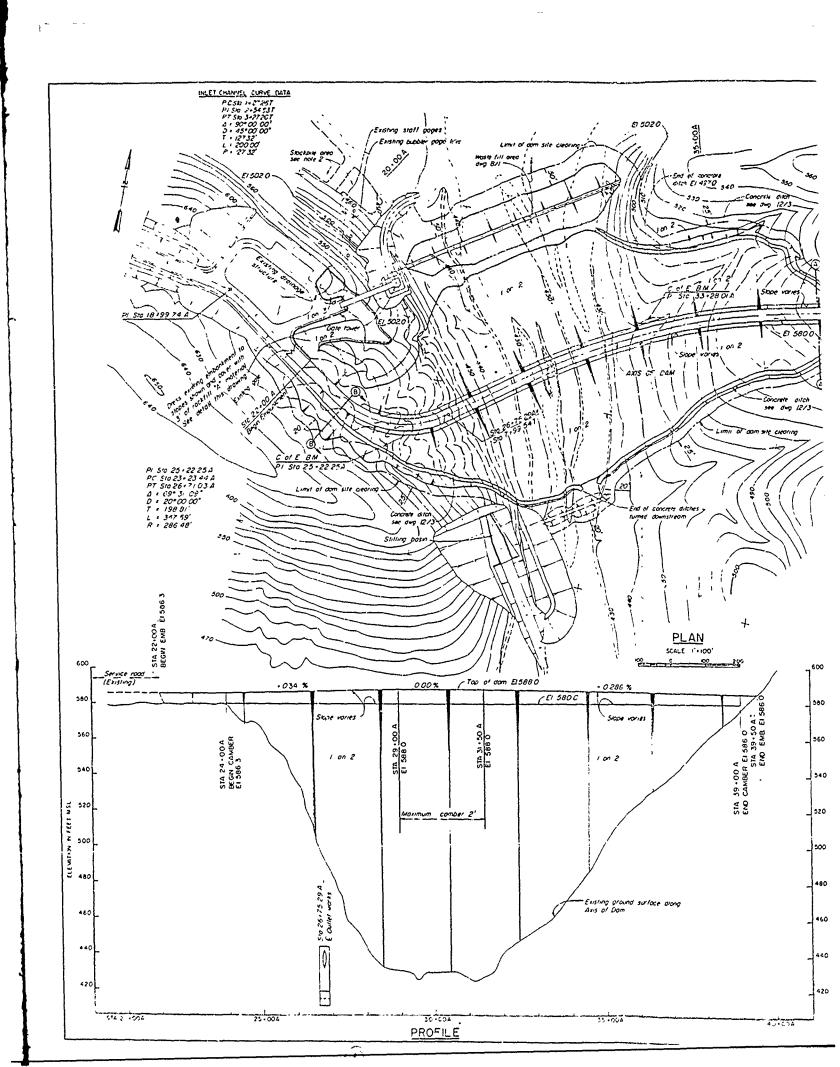


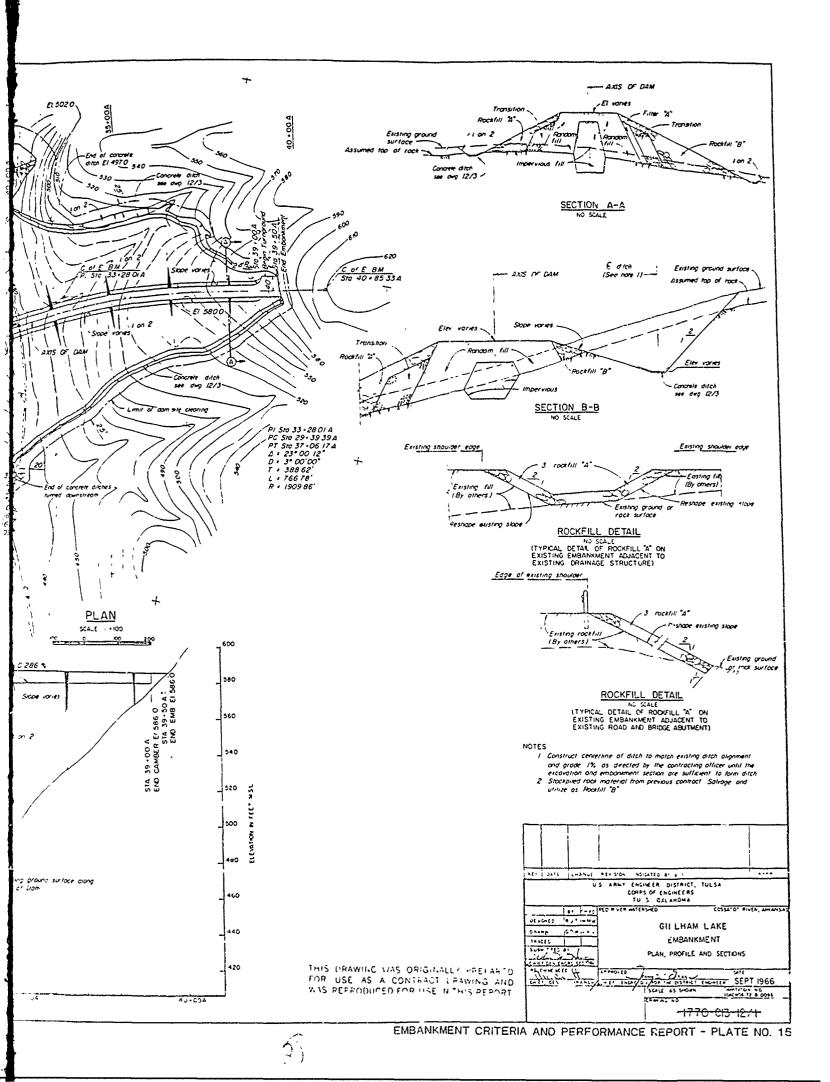


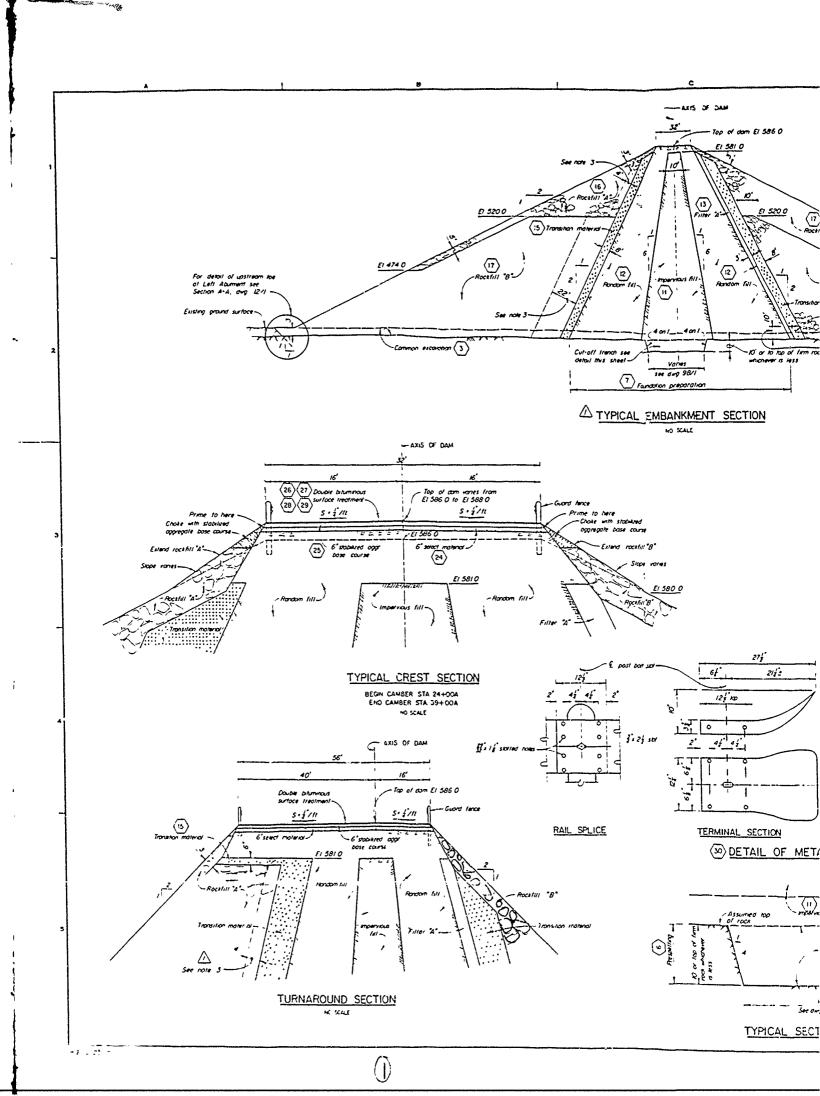


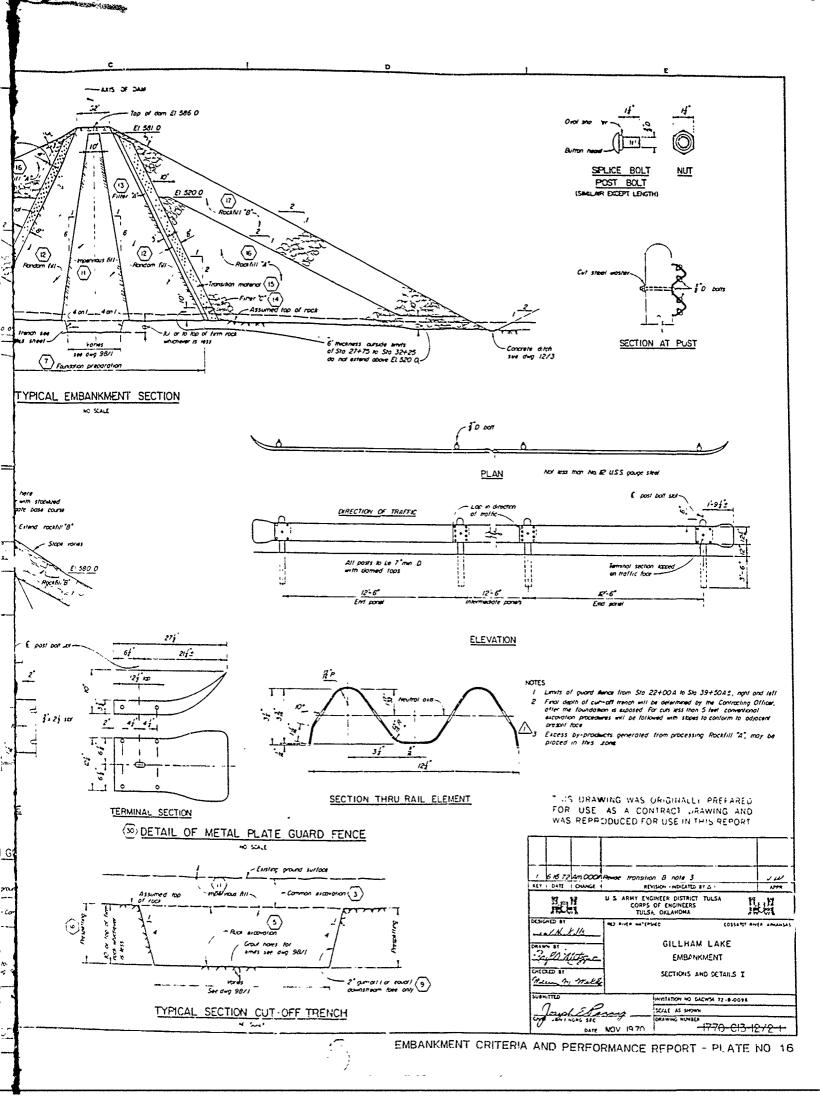


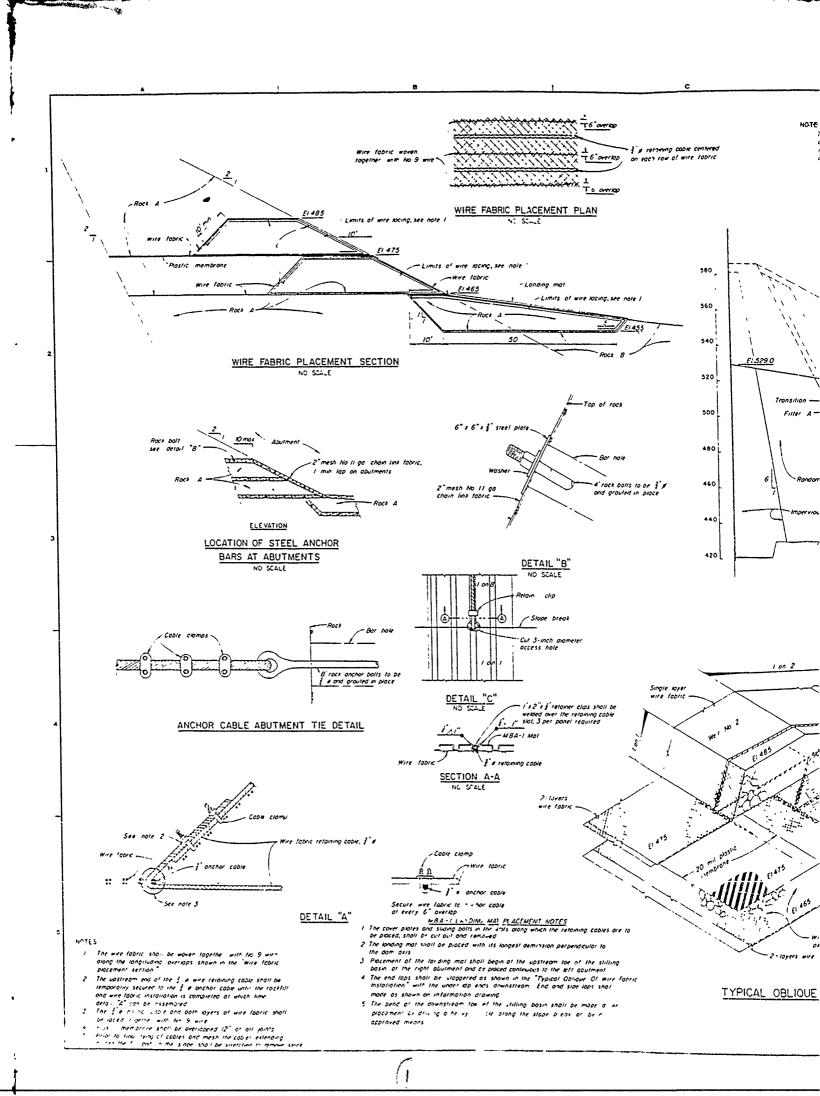


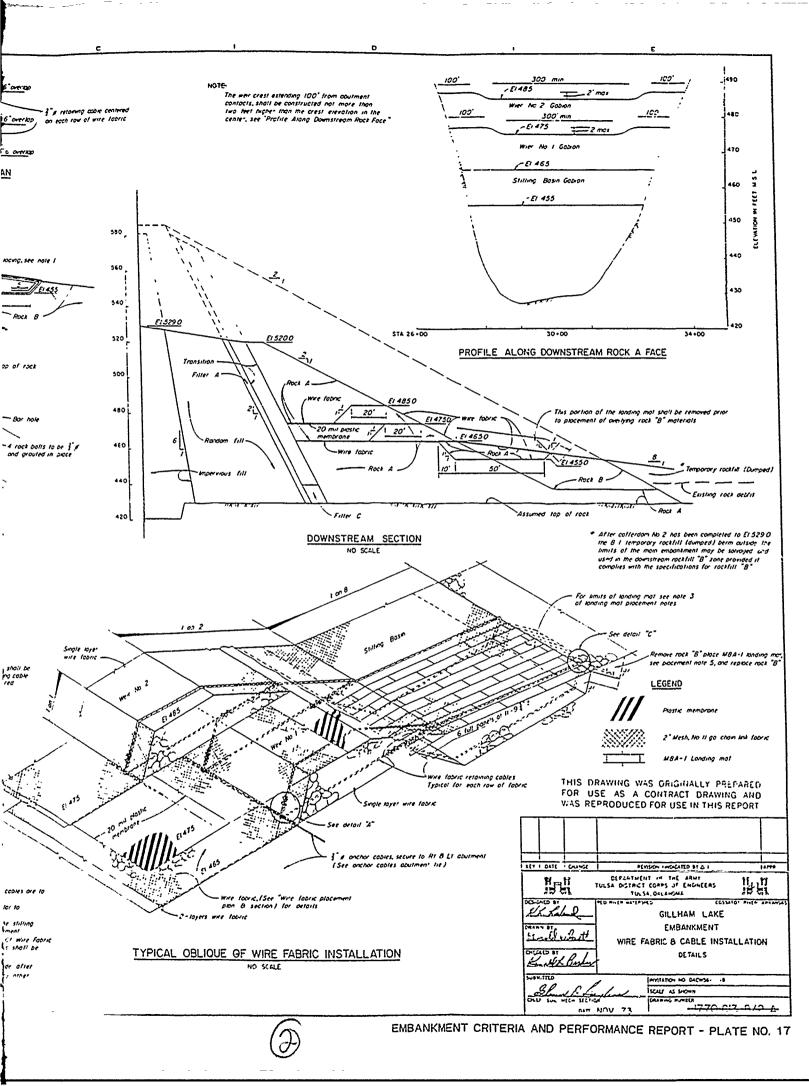


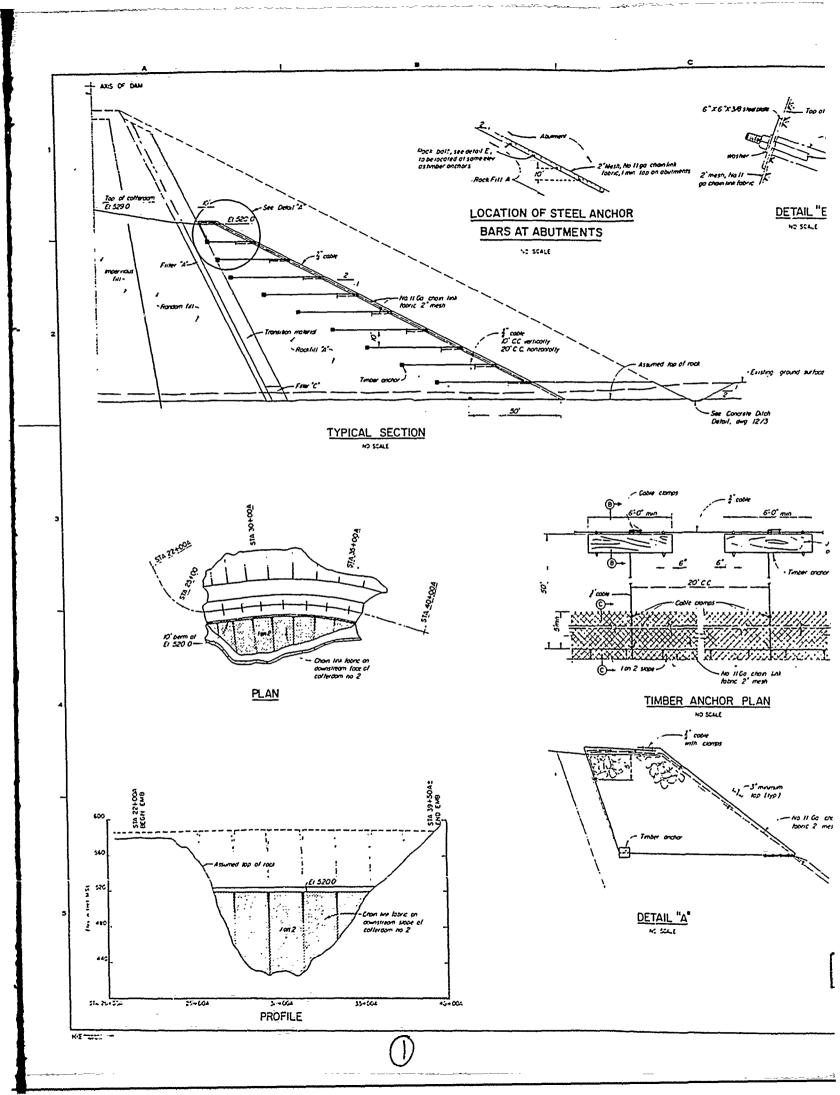


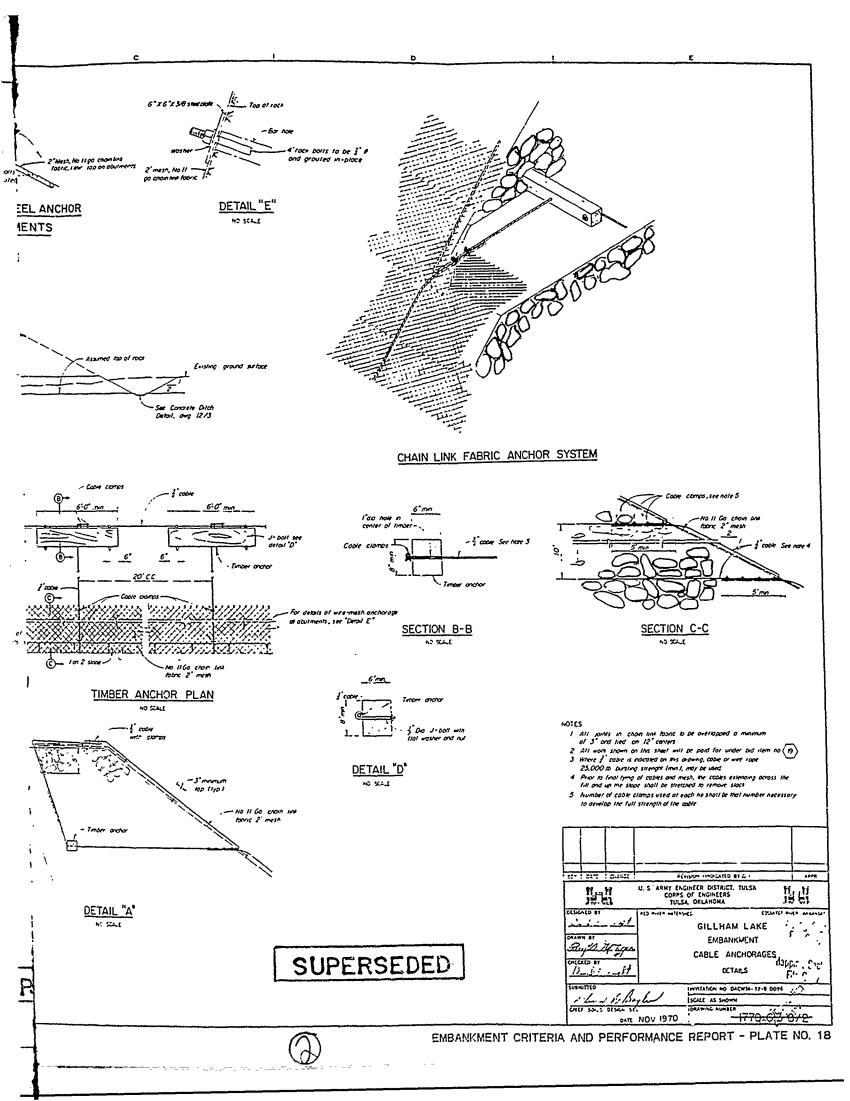


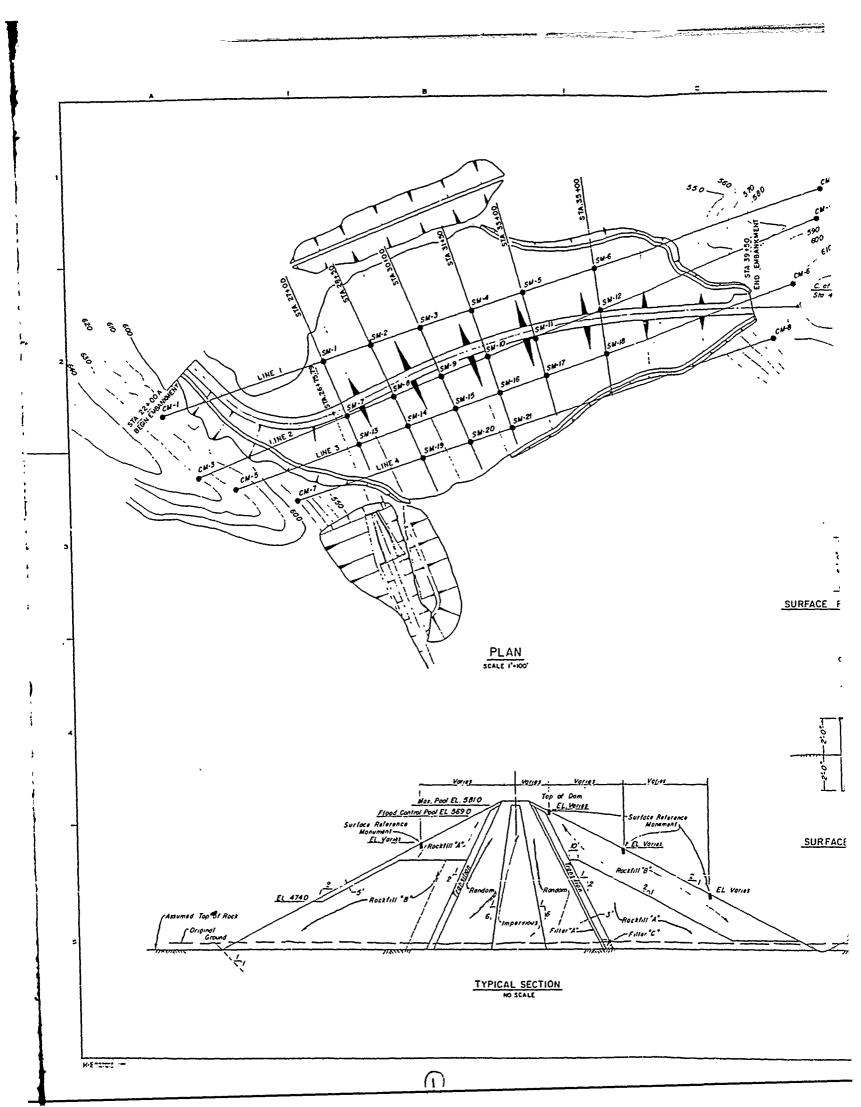


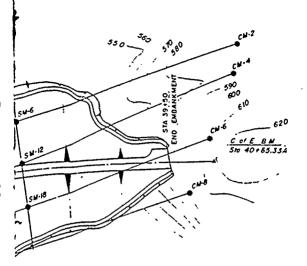












<u>E A</u>

TN

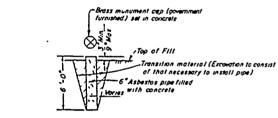
EL Vories

-Filler *C*

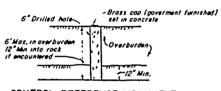
EL Varies

HUBNO	STATION	OFFSET	KUB NO.	STATION	OFFSET
LINE 1			LINE 3		
CM-I	STA.22+45	170'D S	CM-5	STA 24+50	216' DS
SM-I	STA 27 +00	142'U S	SM-13	STA,27+00	110, 02
SM-2	STA 28 4 50	124'US	SM-14	STA 28+50	126' 05
SM-3	STA 30+00	108,02	SM-15	STA.30+00	136, D2
SM-4	STA 31+50	101 U S	SM-16	STA, 31+50	141' D S
SM-5	STA 33+00	109,08	SM-17	STA. 33+00	131, 02
SM - 6	STA. 35+00	136'US	SM-18	STA. 35+00	101, 02
CM - 2	STA, 41+55	335'US	CM-6	STA. 40+70	72' U S
LINE 2			LINE 4		1
CM-3	STA, 23+85	235'05	CM-7	STA, 25+50	216, 0 2
SM-7	STA 27+00	24' DS	SM-19	STA.28+50	224'D S
SM-B	STA 28+50	32, 02	SM - 20	STA 30+00	238 D S
SM-9	STA.30+00	38' DS	SM - 21	STA.31+50	246 D S
SM-10		36' DS	CM-8	STA.40+10	40' DS
SM-II	STA.33+CO	20, D2	1	i	1
SM-12	STA 35+00	20 US	1	i	1
CM-4	STA 41+40	255'05	1	1	1

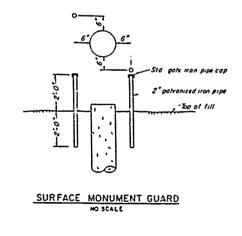
Locations of central monuments and offsets upstream and downstream of the dam axis for surface manuments are appreximate, Exact locations to be determined in the field.



SUP LCE REFERENCE MONUMENT MO SCALE



CONTROL REFERENCE MONUMENT



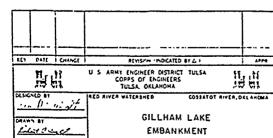
LEGEND

- SM Surface reference monument
- CM Control reference manument

NOTE

i All work shown on this arowing will be paid for under bid them (20)

THIS DRAWING WAS URIGINALLY PRETABELY FOR USE AS A CONTRACT DRAWING AND WAS REPRODUCED FOR USE II. THIS PEPCRY

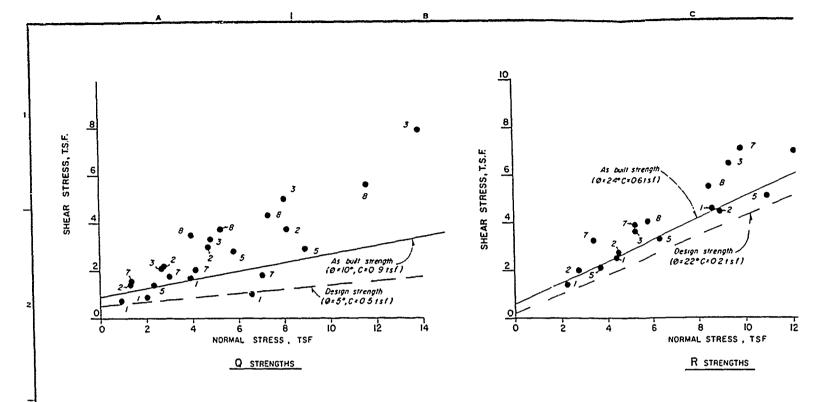


CHECKED BY ENGINEERING MEASUREMENT DEVICES PLAN, SECTION AND DETAILS

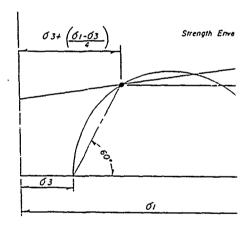
DATE NOV 1970

INVITATION NO DACWS6-72 -8-0096 SCALE AS SHOWN -1770-013-13/1-

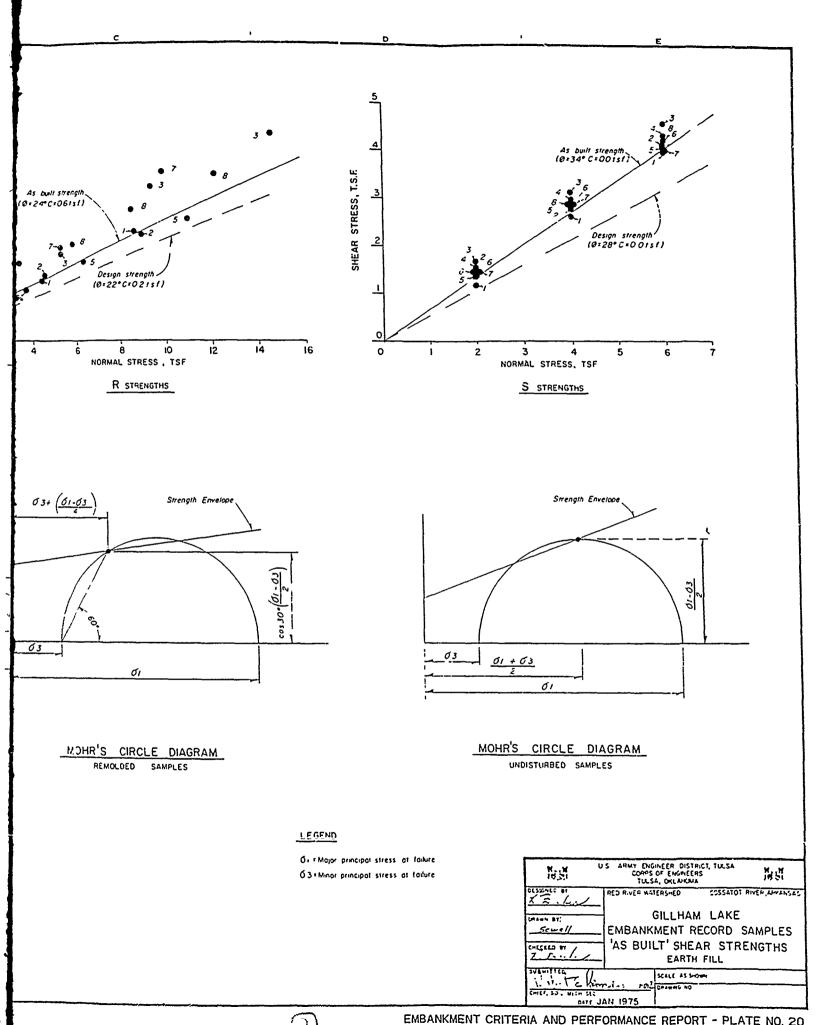
EMBANKMENT CRITERIA AND PERFORMANCE REPORT - PLATE NO. 19

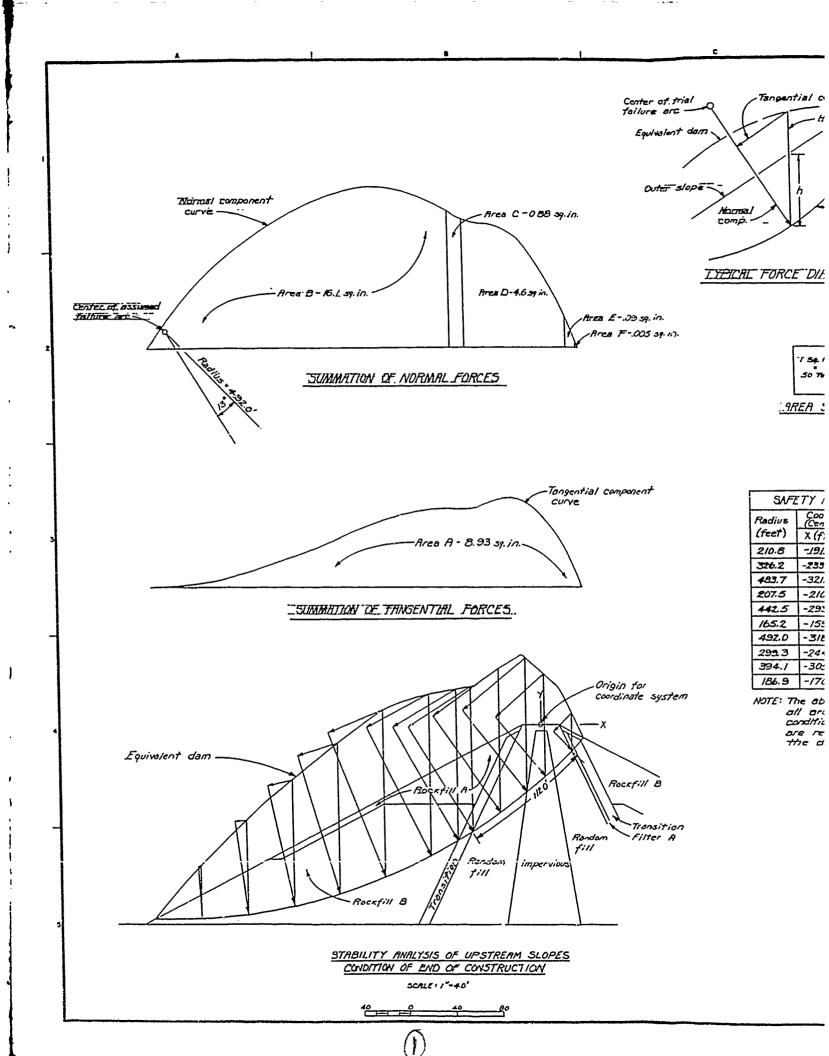


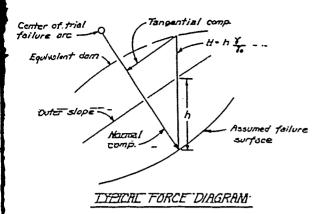
	EMBANKMENT EARTH FILL SAMPLES									
SAMPLE NO	RECORD SAMPLE NUMBER	TYPE SPECIMEN	STA.	OFFSET ft.	ELEV. ft M.S.L.	CLASS	LL %	%Fi.	wc %	Yd p.c.f.
1	1-3	REMOLDED	31+00	Ę	440	CL	23	60	16	113 4
2	R-5	REMOLDED	31+00	60' DS	438	CL-ML	22	49	138	1168
3	R-23	UNDIST	30+00	50' US	460	CL-ML	18	57	HA	115 1
4	*4	UNDIST	31+00	E	480	sc	27	48	147	116
5	* 5	UNDIST	32+00	50'DS	480	sc	22	43	134	118
6	#6	REMOLDED	28+70	45'US	504	sc	21	41	151	121
7	1-44A	REMOLDED	30+00	£	532	CL	22	65	142	119
8	#8	UNDIST	35+00	15'05	550	CL	29	79	171	115 7



MOHR'S CIRCLE DIAGRAM
REMOLDED SAMPLES







F-.09 sq in. F-.005 sq.in.

T SA IN.
SO THIS

mpanent

is for

in for Unate system

> Transition Filter A

cefill B

SAFETY FACTOR SUMMARY								
Radius	Coordir Center	nates	Sofety					
(feet)	x (ft)	Y (5t)	factor					
210.8	-1910	4Z.0	1.759					
326.2	-235.5	161.0	1.582					
483.7	-321.D	324.0	1.564					
207.5	-210.0	42.0	1.574					
442.5	-293.0	281.0	1.560					
165.2	-159.5	20.0	1.854					
49z.0	-3/8.0	329.0	1.555					
293.3	-244.5	1465	1.663					
394./	-305.0	228.5	1.627					
186.9	-170.0	22.0	1.791					

NOTE: The above table includes results of all arcs analyzed for end of construction condition. Coordinates for arc centers are referenced from & and crest of the dam.

ADOPTED DESIGN DATA								
MATERIALS	Soil wts. L	bs /F+3	Shear strengths					
MITTENIALS	Saturated	Submerged	*	Q	R	5		
ROCKFILL A	105 **	67.6	φ	42	4Z	42		
7,007,722, 77	/30	00	С	0	0	0		
ROCKFILL B	/3 <i>5</i>	72.6	ø	36	.36	36		
NOCIVILL D	733		С	0	O	0		
TRANSITION	125	62.5	ø	35	33	33		
77 0 17 10 77 7 10 1			C	0	a	0		
FILTER A	125	62.5	Ð	33	33	33		
FILIEN A	125	α2.5	C	0	0	0		
FRNDOM FILL	125	62.5	Ø	5	22	28		
TITUIDUM TILL	125	62.3	С	0.5	0.2	0		
IMPERVIOUS	125	62.5	ø	5	22	28		
WIII- ENVIOUS	/25	12.5	C	0.5	az	Ω		

* # Angle of Internal friction (Degrees)

* * Seturated surface dry weight (40% Yords)

C * Cohesion (Tons/ff*)

Forces acting on failure or

Positive tangential force-Area A = 4465 Tons ETotal • 4465 Tons

Normal and resisting forces

Normal force Area B	=805 Tons
Normal force Area C	· 440 Tons
Normal force fires D	* 230.0 Tans
Normal force Area E	. 4.5 Tons
Normal force Area F	- 0.25 Tans
Cohesion (LC)= 1/2.0 × 0.5	= 56.0 Tons

Safety factor computations

Formula: S.F. - ENxtan++LC ET

For unconsolidated-undrained (Q) condition

5.F. = 805.75 tañ 36° + 48.5 tan 33° + 230 tan 5° + 56.0
.3446.5

S.F. = 1.55.

	No.			CORP: TUL	GINEER DISTRICT, S OF ENGINEERS SA, OKLAHOMA		
	TRACED SUMMITTED	MIG MIG	SL.	ST. OF	IAM DAM AN EMBANK ABILITY ANA CONSTRUCTI	ID RES MENT LYSIS-	END
۲.,		w		STATE COME OF	SCALE, AS SHOWN	hing (n	FEB. 1971
						DM9	-9 8/7

ADC	PTED D	ESIGN D	MTA			
EFFT DIW'E	-501L-WTS.	185/FT3	SHE	AR 57	REN67	ገዣ -
MATTERIAL'S	SATURATED	SUBMICRSED	*-	Q	· R	5
	105.**	~7.6	ø	42	42	42
ROCKFILL.IA	730	67.6	C	0	Ö.	.Ø
THINKELL D	,,-	,	•\$	36	<i>3</i> 6	34
ROCKFILL B	CKFILL B 135 72.6	C	G	Ø	0	
TRANSFION	-125	62.5	ø	33	33	:3:3:
איסו ועבומתיו ב.	-725		Ċ	0	0	.D
5"750 A	200	10.5	Ø	33	33	33
FILTER A	125.	62.5	C	.0	Ø	_2
RPANIDOM FILL	125	EQ E	Ø	.5	22	28
THOUSAN FILL	123	62.5	·c	a.5	0:2	·a
BADEDWAR	125	62:5	ø	5	22	28
MPERYIOUS	123	02.5	C	0.5	0.2	:0.

表す。Angle of internal friction (Degrees) 表式で Saturated surface dry weight (40% Voids) で で Cohesion (Tons/ft²)

15 ACTING ON FAILURE ARC.

cossitive tangential force-firea A = 522 Tons Negative tangential force-firea B =-16 Tons ETotal = 506 Tons

MORMAL AND RESISTING FORCES

Cobesion (LC) = 148 x 0.5 • 74 Tons

SMEETY FACTOR COMPUTATIONS _.

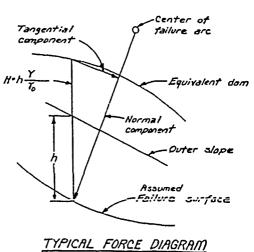
Formula To SEE ENT for & FEE

For inconsolidated undrained (Q) condition

5.74 BDS tan 42" + 83.5 tan 33" 3.4 07 tan 5" + 2.5 tan 36" + .74

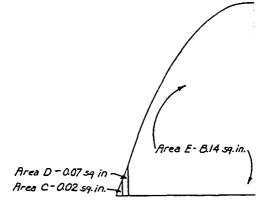
S.F; - F.775

SAFE	TY FACT	OR SUM	MARY
Radius (Fæt)	Coordii (Center . X.(Ft.)	of Arc)	Safety Factor
210.8	.191.0	47.0	1.9/5
926.2	233.5	161.0	1.774
.±63.7	32/.0.	324.D	1:805
Z0Z5	210.0	.42.0	2095
.442.5	295.0	281.0	1.794
165.2	J59.5	20.0	2,004
4920	9/8.0	325.0	1.798
299.3	244.5	146.5	1.074
.394.1	305.0	228:5	LB84
106.9	17.2.0	: 22.0	1.913

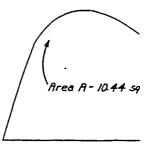


NOTE:

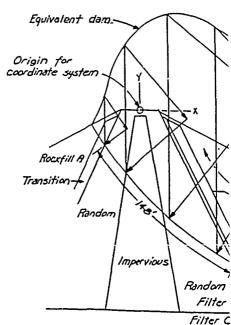
The above table includes results of all arcs conjuged for the and of construction condition. Consideration for an intersacial from the dame.



SUMMAT

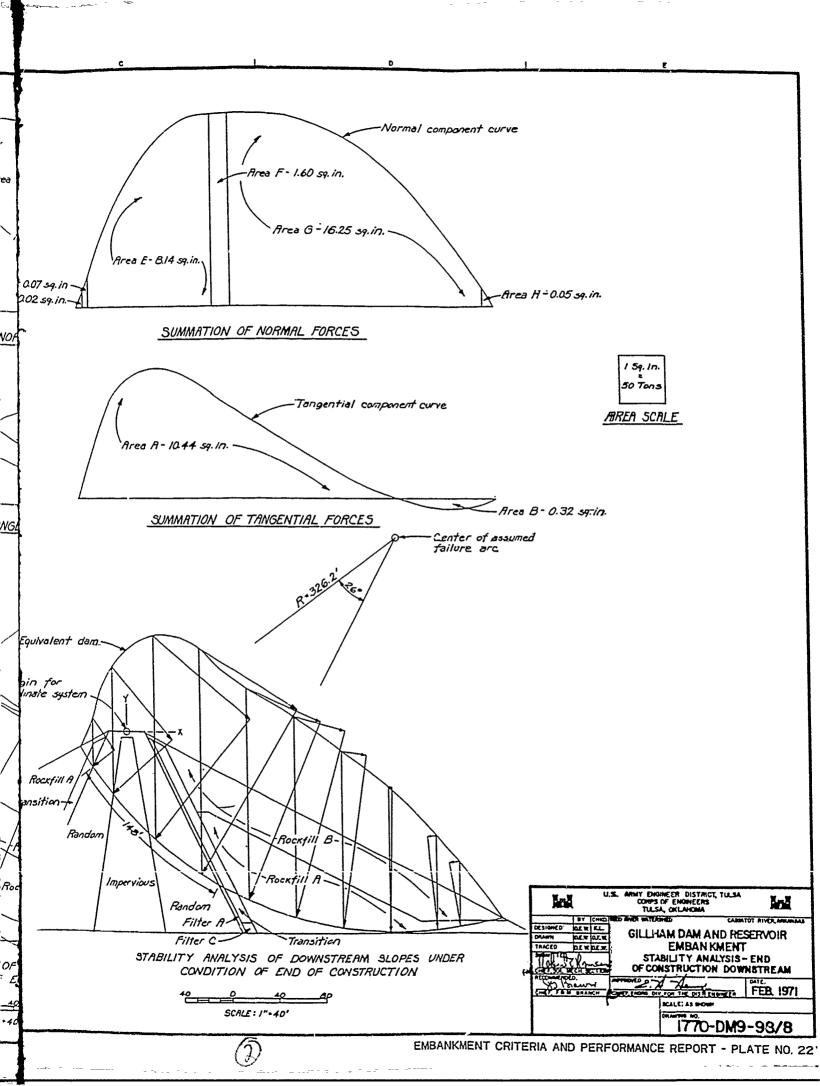


<u>SUMMATIOI</u>



STABILITY AN

40



A	DOPTED	DESIGN	VALL	IES		
INATERIAL C	SOL WTS.	185/FT5	SHE	AR S	TRENC	THS.
MATERIALS	Saturated	Submerged	*	Q	R	s
<i></i>	105**	67.6	ø	42	42	42
Ploskfil! A	150	07.0	C	0	0	0
Rockfull B	135	72.6	ø	36	36	36
MUCKIIIL D	755		C	0	0	0
Transition	125	62.5	d	33	33	33
//			C	0	0	0
Filter A	125	62.5	1	55	55	35
	125	02.5	C	0	0	0
Random Fill	125	62.5		5	22	28
, (8) (100 p) F) (1	125 82.5	02.5	C	0.5	0.2	0
/	100	40.5	0	5	22	28
Impervious	125 62.5	6	0.5	0.2	0	

* Angle of Internal Priction (degree)

* Saturated Surface Dry weight (40% Voids)

C * Cohesion (Tans (FE2)

150 IN.

30 IDNS

AREA SCALE

150
Area C=11.92 SQ.INL

Area C=0.02 SQ.INL

Area D=

106 SQ.NL

SUMMATION OF NORMAL.

FARCES ACTING ON FAILURE ARC Porture longantial force-Area A= 549.5 tons Negative longential force Area B= -64.0 tons

E TOTAL = 485.5 tons

NORMAL AND RESI Normal force Hree C Normal force Area D AND RESISTING FORCES 7705 3.0 Tons Normal force Area E 265.0 Tons Normal force Area F 75:0 Tons Normal force Area G 596.0 Tons Normal Force Area H 2.0 Tons SAFETY FACTOR COMPUTATIONS

FORMULE: S.F. ENX ton \$+LC

.. _ For consolidated-Urained (3) condition

SF = 597 (an 42°+ 78 ton 35°+ 265.0 (an 28°+ 2.0 (an 35°
485.5

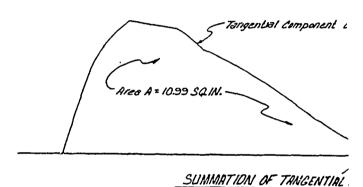
SF = 1.50

For consolidated-undrained (R) condition

5.F. • 597 ton 42° + 78 ton 33° + 265.0 ton 22° + 2 ton 36° + 29 5

SF. = <u>1.495</u>

Average Rand 5 . 1.50



Equivalent dam

Equivalent dam

Stan36+295

Origin for Coordinate system

Flood control pool EL. 569.0

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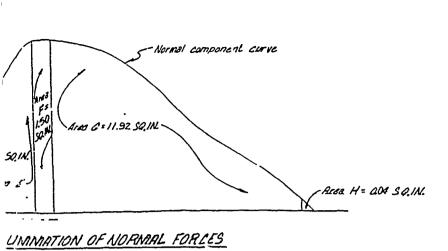
Rockfill S

Rockfill S

STABILITY ANALYSIS OF DOWNSTREAM SLOPE

<u>CONDITION OF STEADY SEEPAGE</u>

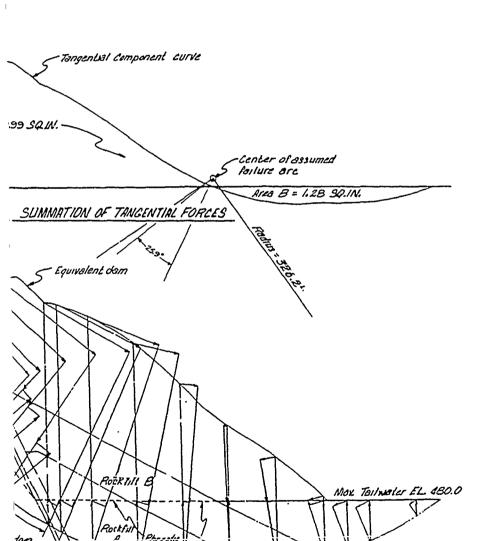
SCALE: 1"-40'



-Center of assumed failure are Tangential component Equivalent dam component -Outer slope - Saturation line failure surface

TYPICAL FORCE DIAGRAM

NIHI	ION	UF	ישעע	UVINL	FUNL	رع



S	SAFETY FACTOR SUMMARY								
RADIUS	Coordu Center o	PALES	Sofely i	Factor					
(FT.)	X (fl)	Y(ft)	Rstrongth	Sstrength					
210.8	191.0	97.0	1.519	1.486					
326.2	233.5	161.0	1.486	1.4.97					
483.7	321.0	324.0	1.595	1.518					
207.5	210.0	42.0	1.638	1.593					
442.5	293.0	281.0	1.557	1.551					
105.2	/.59.5	20.0	1.642	1.597					
	318.0								
299.3	244.5	146.5	1.575	1.542					
394.1.	305.0	228.5	1.615	1.585					
186.9	170.0	22.0	1.533	1.505					

Note:

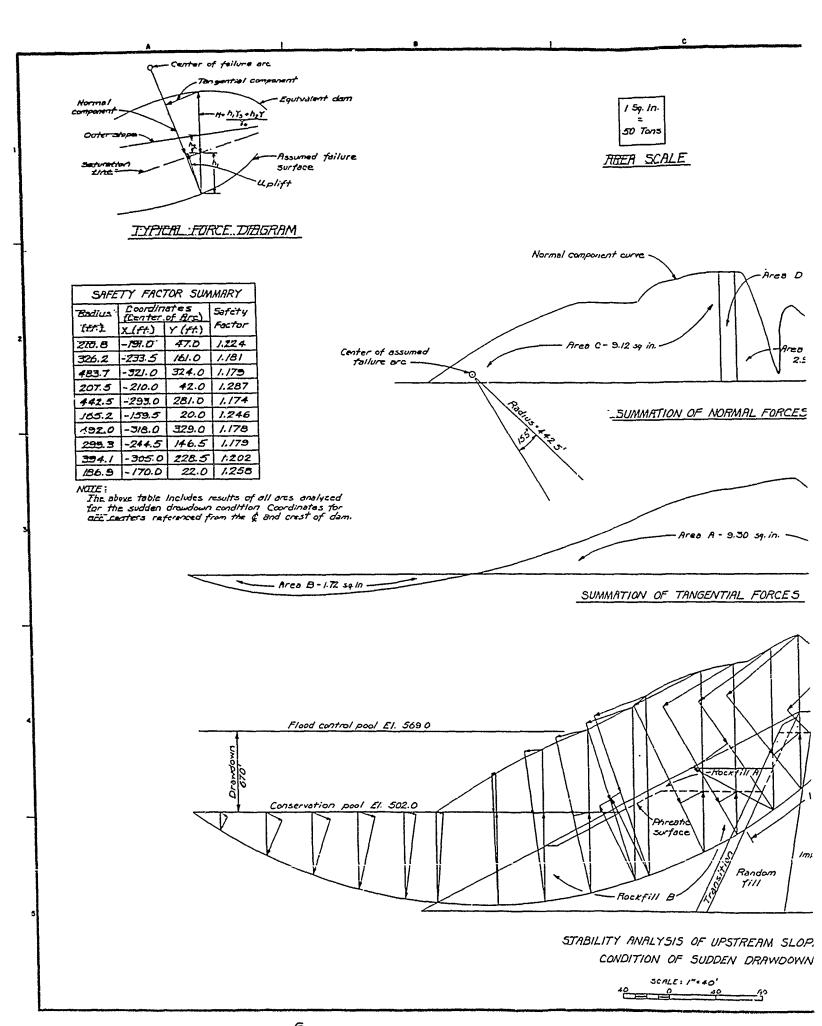
The above table includes results of all arcs analysed for the steady suppose condition coordinates for arc centers are reterented from the Land Crest of the

IS OF DOWNSTREAM	SLOPES	UNDER
I OF STEADY SEEP	AGE	
SCALE: 1" -40'		

Fill

Surtone

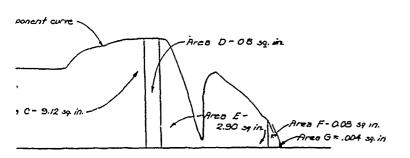
ĀOĀ		US		NEER DISTRIC OF ENGNEERS OKLAHOMA	T TULEA	Acid
	87 (нион	ED RIVER WITERS	7.0	CAMBAT	OT RIVER, ARKHINAS
DESIGNED	DEW	KL	CHLU	M DAM A	אות סבינ	CED/N/CE
DRAWN,	OF W	TEW.	GLLD			
TRACED	DEMI	w 3c		EMBANI	(MENT	
		治	6TABILI1	Y ANALYSI	S-STEAL	DY SEEPAGE
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				1770	-DM9	- 98/9



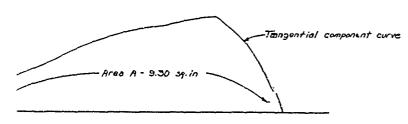
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1 59. In. 50 Tons

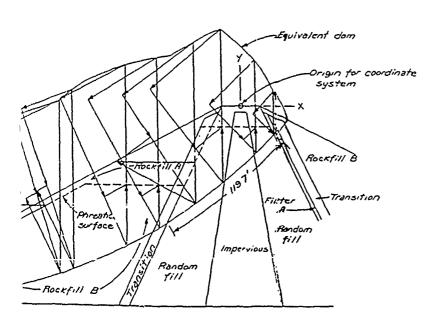
THER SCALE



SUMMATION OF NORMAL FORCES_



JMMATION OF TANGENTIAL FORCES



LITY ANALYSIS OF UPSTREAM SLOPES UNDER CONDITION OF SUDDEN DRAWDOWN

SCALE: /**40'

ADOPTED DESIGN DATA							
MATERIALS	Soil Wts. 1	Shear Strength's					
	5eturated	Subneged	* .	Q	R	5	
ROCKFILL A	105** 130	67.6	ø	42	47	42	
		07.0	u	0	æ	0	
ROCKFILL B	/35	72.6	ø	36	36	<i>5</i> 6	
			C	0	9	0	
TRANSITION	125	62.5	Ø	33	Ħ	33	
			C	٥	2	Ю.	
FILTER A.	/25	62.5	ø	53	3	33	
			C ·	0	2	O	
RANDOM FILL	125	62.5	Ø	5	2	28	
			C	0.5	02	2	
IMPERVIOUS	125	62.5	ø	5	27	28	
			C	0.5	OZ	.0	
* Ø * Angle of internal friction (degrees) ** * Saturated surface dry weight (40% Voils)							

Forces acting on failure arc

C = Cohesion (Tons/F+2)

Positive tangential force-Area A = 465.0 Tans Negative tangential force-Area B = -85.0 Tans

ETotal = 379.0 Tons

Normal and resisting forces

Normal force Area C-- • 456.0 Taris Normal force Area D-- +0.0 Texts Moonel force Rrea E-- = 145.0 Tors Normal force Area F-Cohesion (LC) = 119.7' x 02 * 23 9 . Tons

Safety factor computations

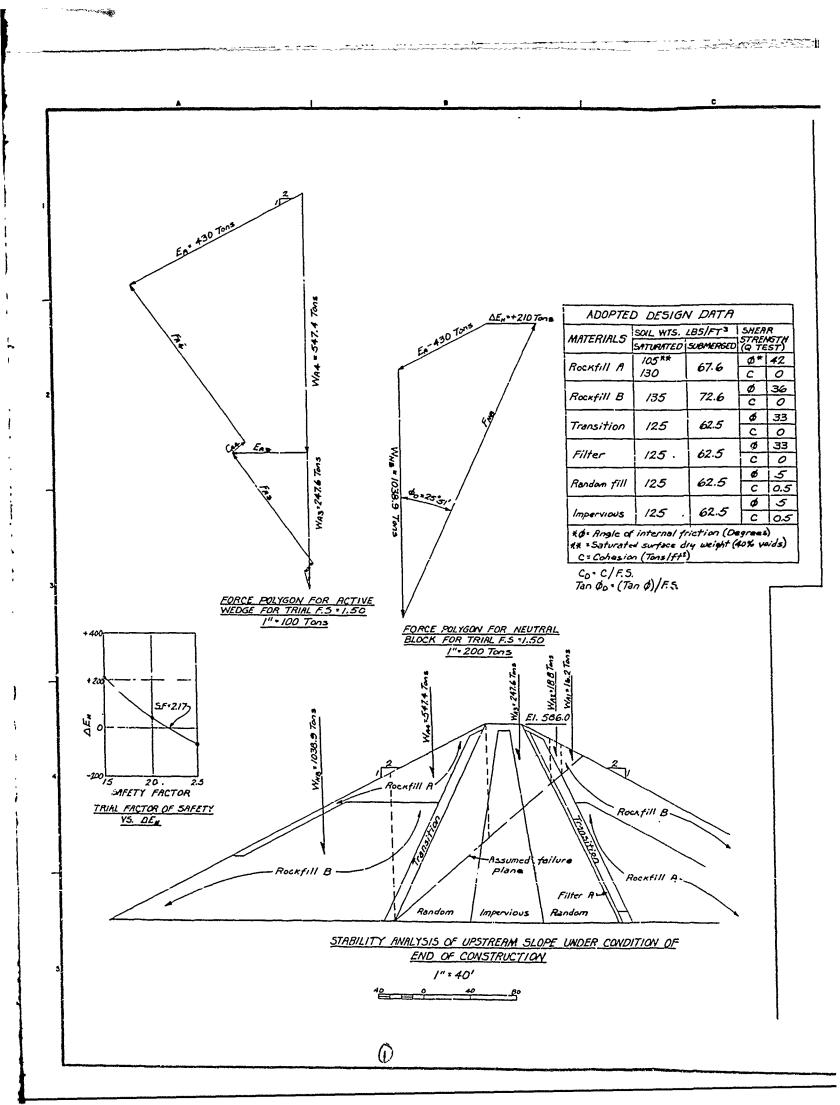
Formula S.F. = ENxtan + LC

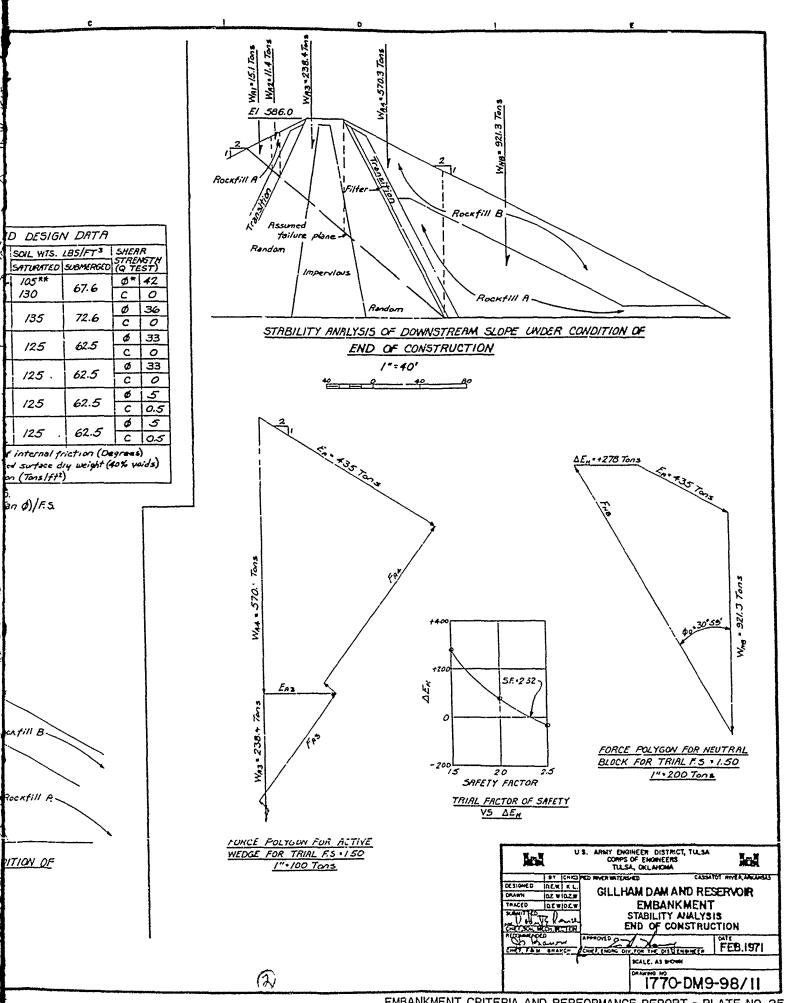
For consolidated undrained (R) condition

5.F. = 456.2 tan 36*+44 tan 33*+1450 tan 77+23.9

5 F. = <u>L.168</u>

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DESIGNED OPANY TRACED SOMETED. LUL TE	OF W	A L A E W A C W	GILLHAM DAM ANDR EMBANK MEN STABILITY ANALYSIS-SIMO	IT
ERE ET BE	nim		CIMEL ENGINE DIS CON THE DIST CHEMELE SCALE: AS SHOWN DRIANTING NO	9-98/10





ADOPTED DESIGN DATA							
MATERIALS	SOIL WTS	SHEAR STREAGTH					
	SATURATED	SUBMERGED	*	R	5		
Rockfill A	105** 130	67.6	Ø	42	42		
			С	0	0		
Rockfill B	/3.5	72.6	Ø	36	36		
			C	0	0		
	125	62.5	Ø	33	33		
Transition			C	0	0		
Filter A	125	62.5	Ø	33	33		
			С	0	0		
Random	125	62.5	Ø	22	28		
			C	0.2	0		
Impervious	125 62.5		Ø	22	28		
		62.5	C	0.2	0		

A 12 Martin Comment of the Comment o

SHEAR STRENGTH TO USE FOR

h = 30.75' (Use R+5/2)

RANDOM AND IMPERVIOUS MATERIAL

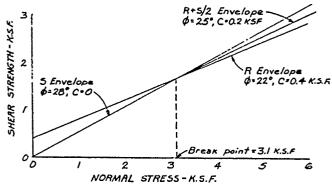
Solve for h for N = 3.1 K.S.F. (See section)
[(586-569)=0.125+0.0625] Cos 40° = 3.1 K.S.F.

* \$\Phi^* Angle of internal friction (Degrees)

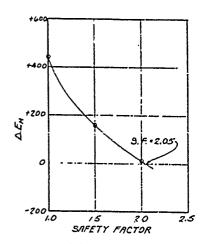
** * Seturated surface dry weight (fo% voids)

C: Cohesion (Tens/Ff2)

 $C_0 = C/F.5.$ Tan $\phi_0 = (Tan \phi/F.5.)$

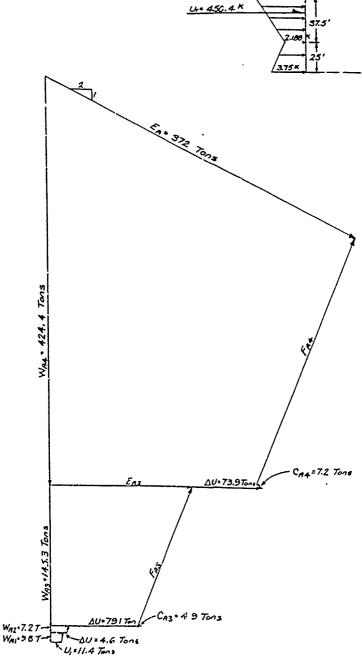


COMPOSITE STRENGTH ENVELOPES FOR RANDOM AND IMPERVIOUS MATERIALS



TRIAL FRCTOR OF SAFETY

VS. ΔΕΗ

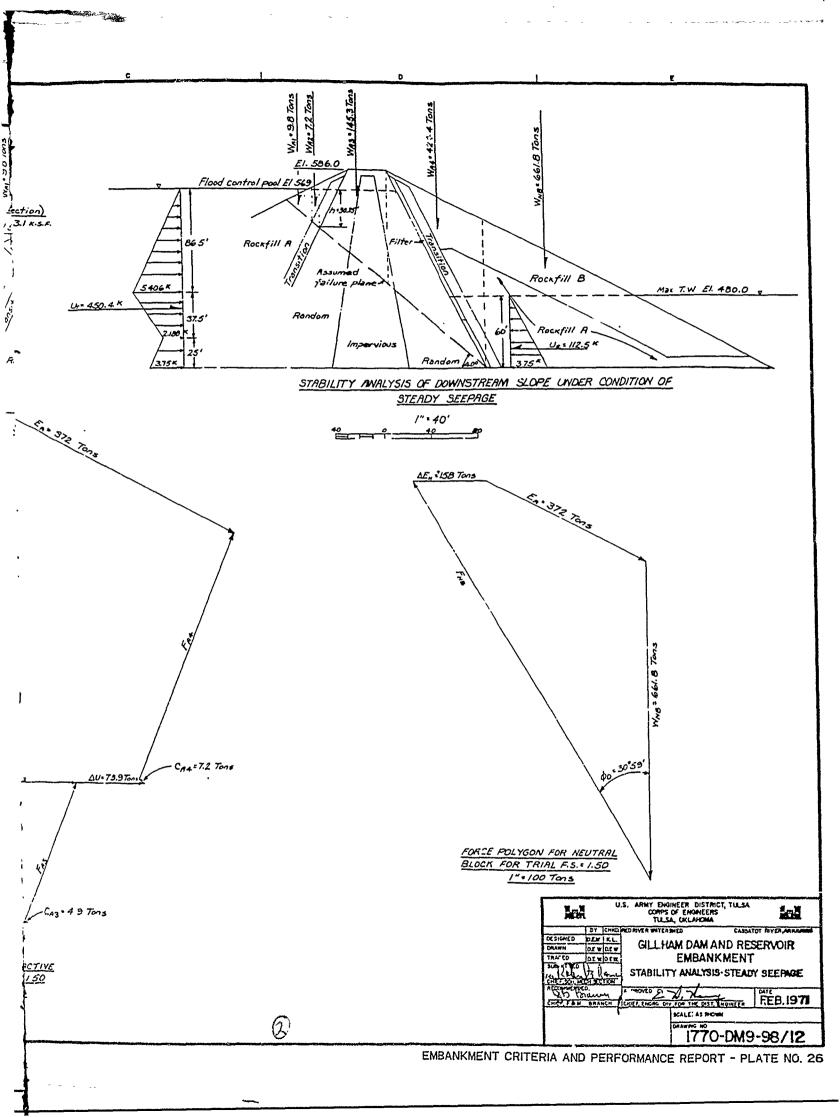


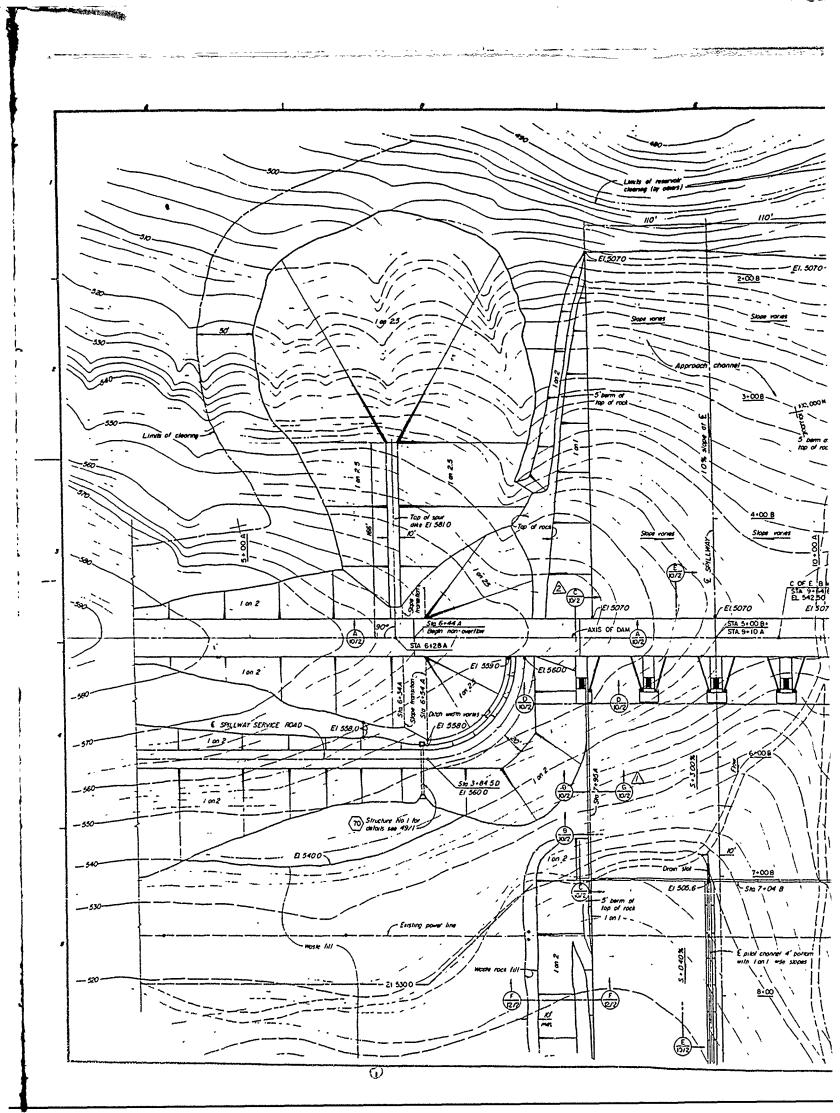
Flood co

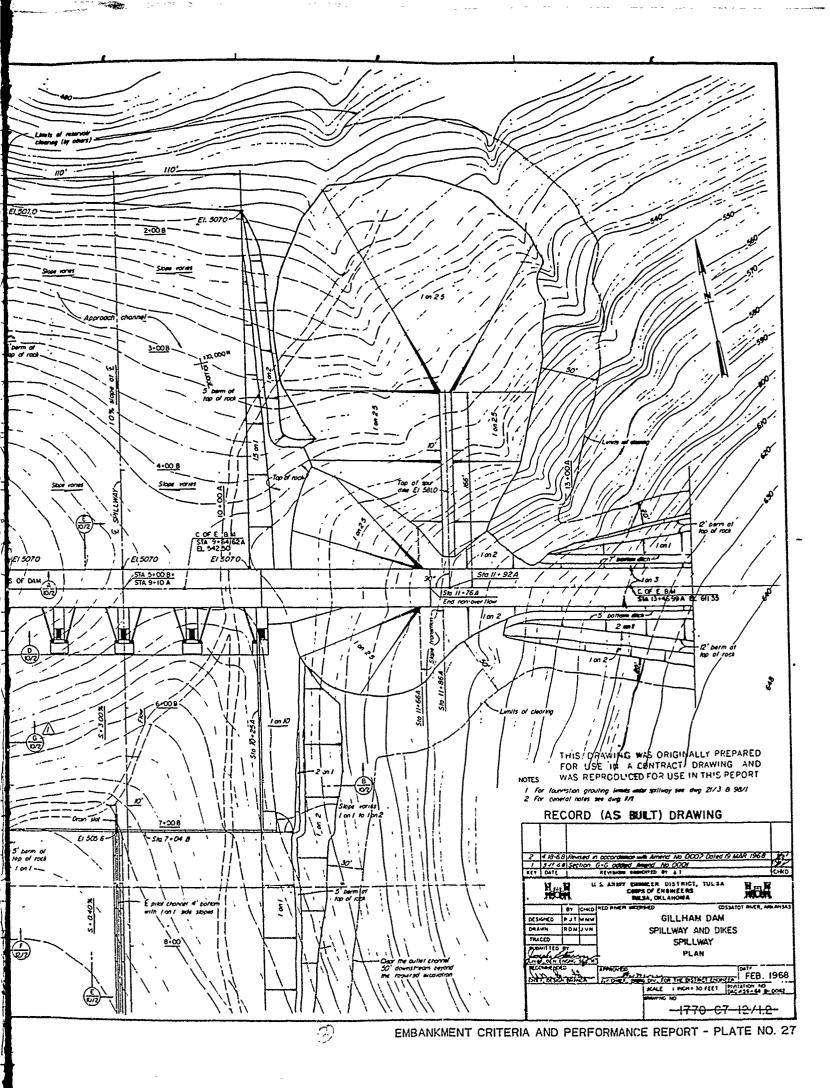
865'

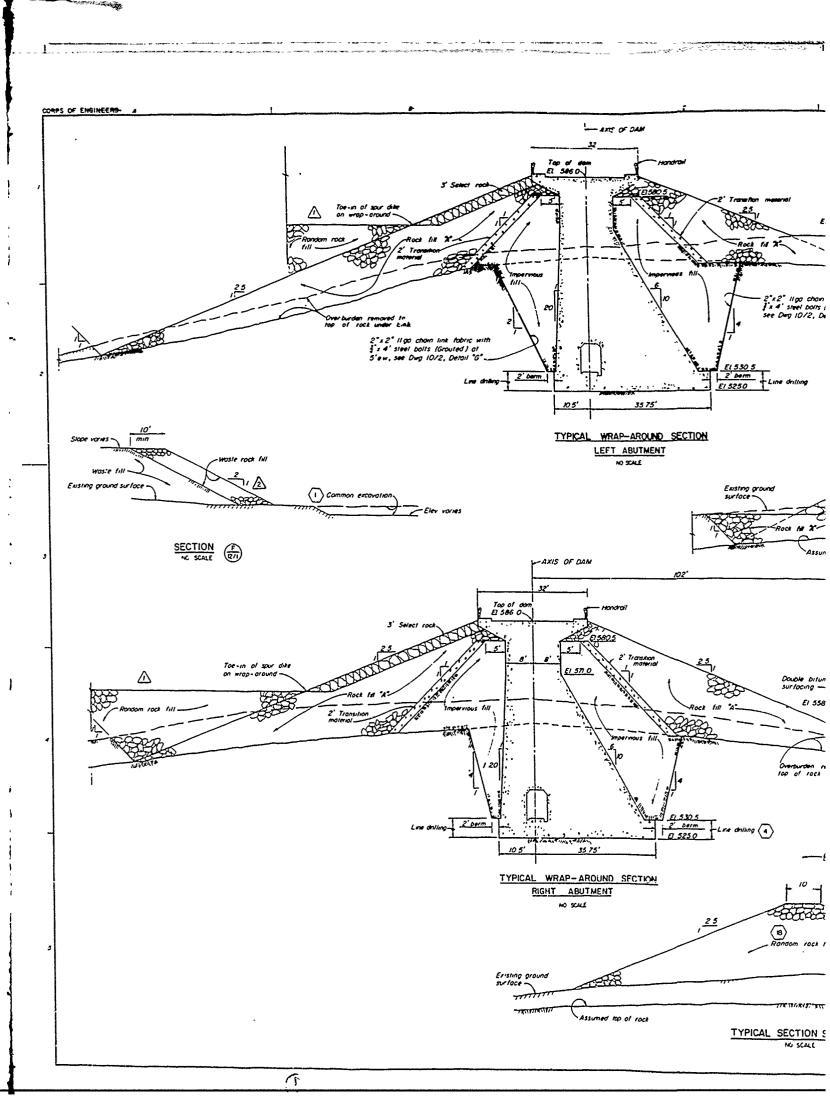
EORCE POLYGON FOR ACTIVE WEDGE FOR TRIAL FS - 150

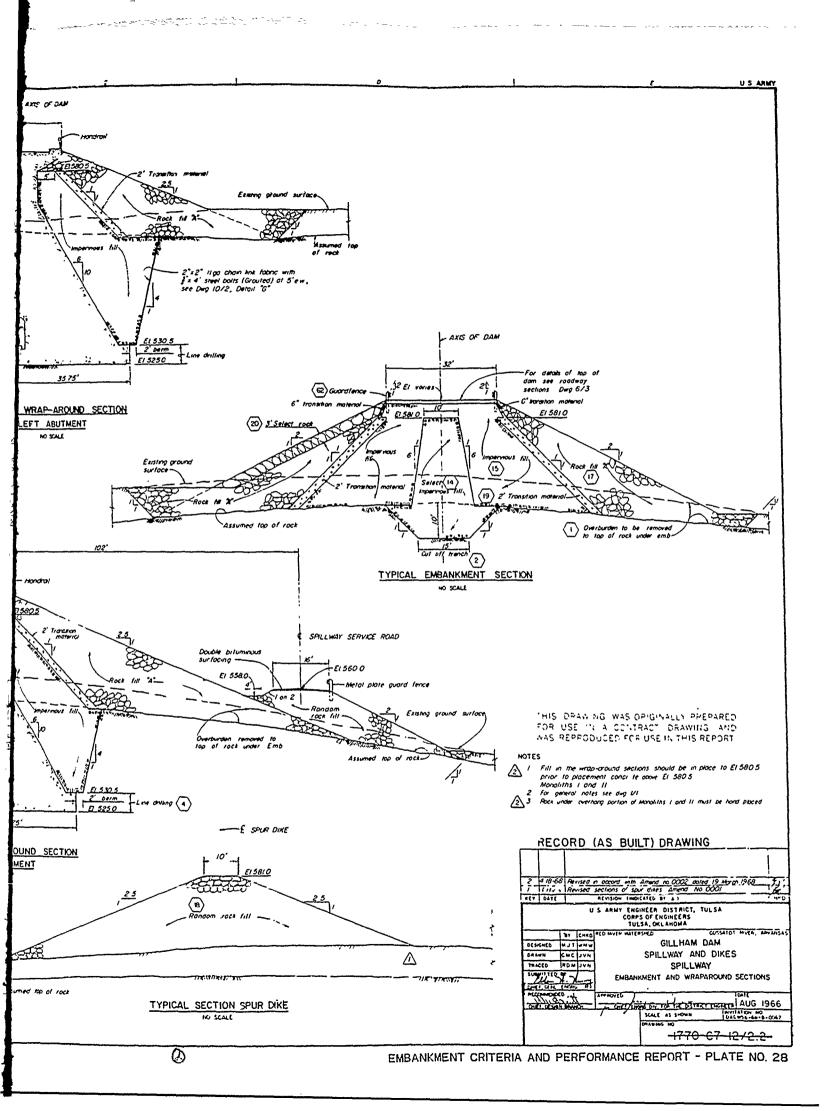
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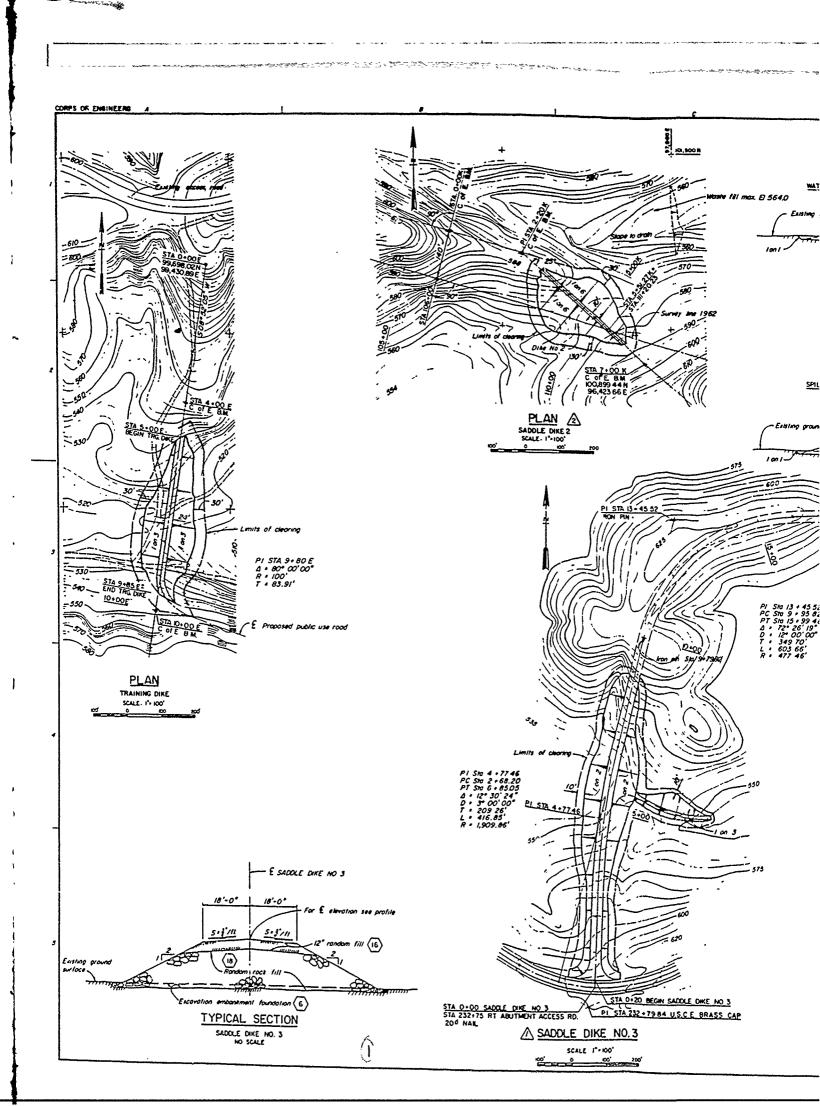


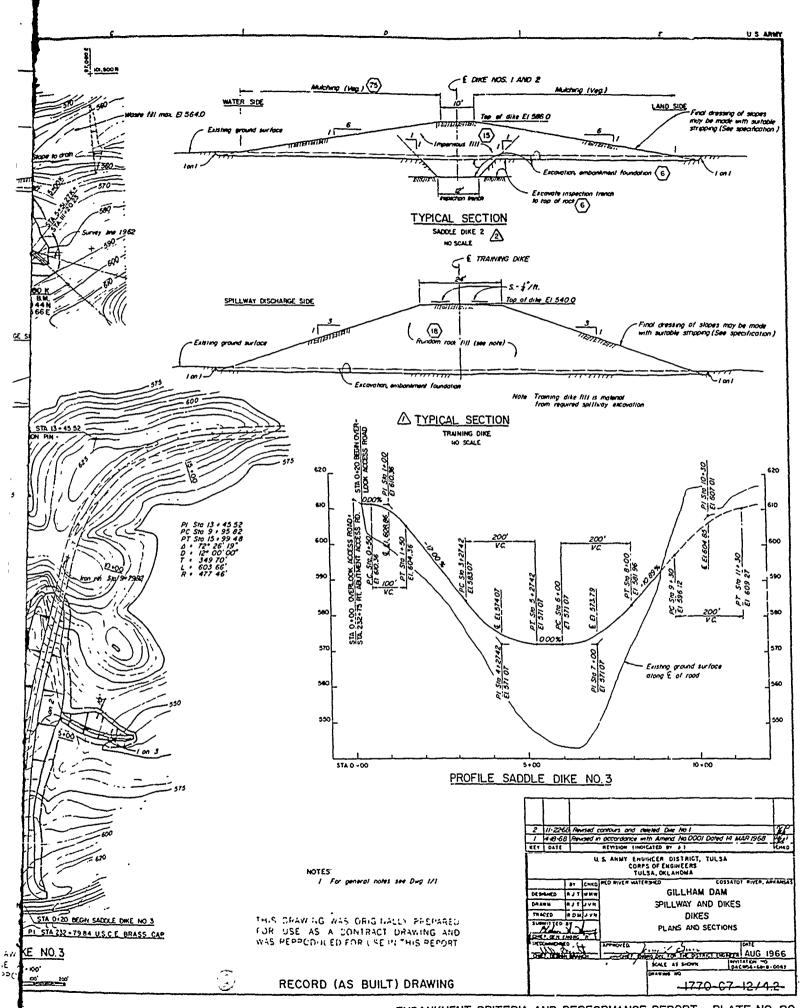






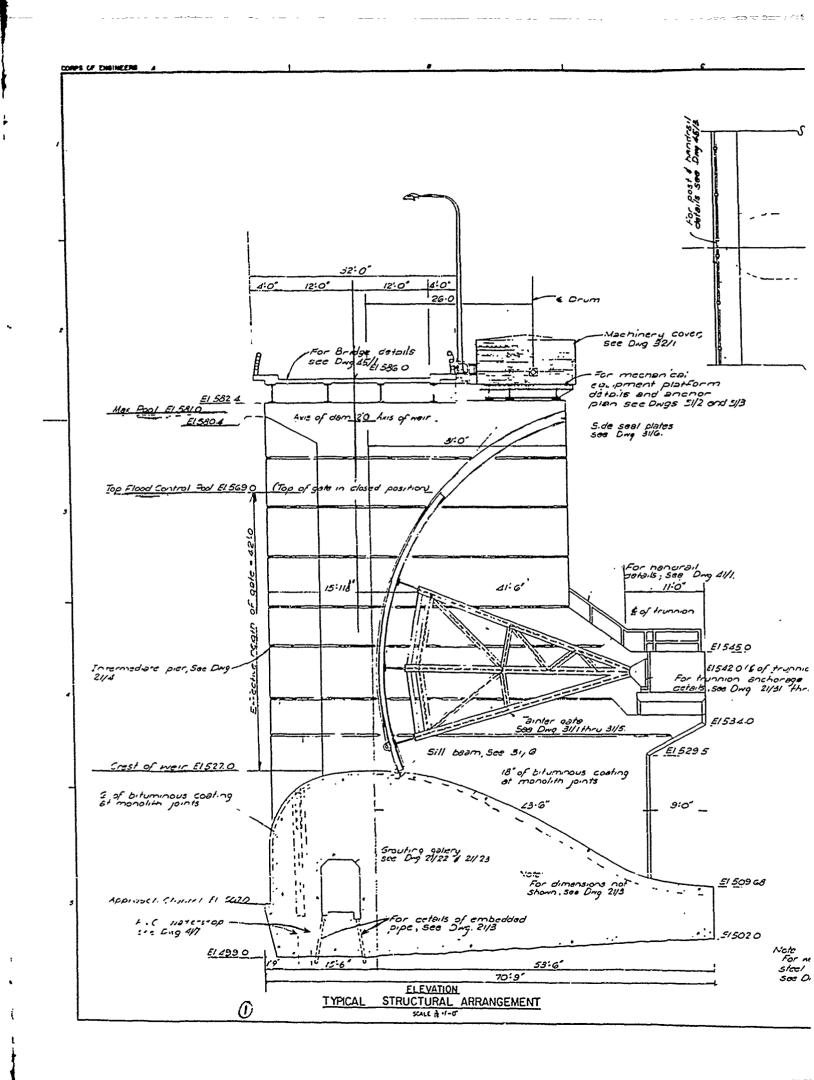


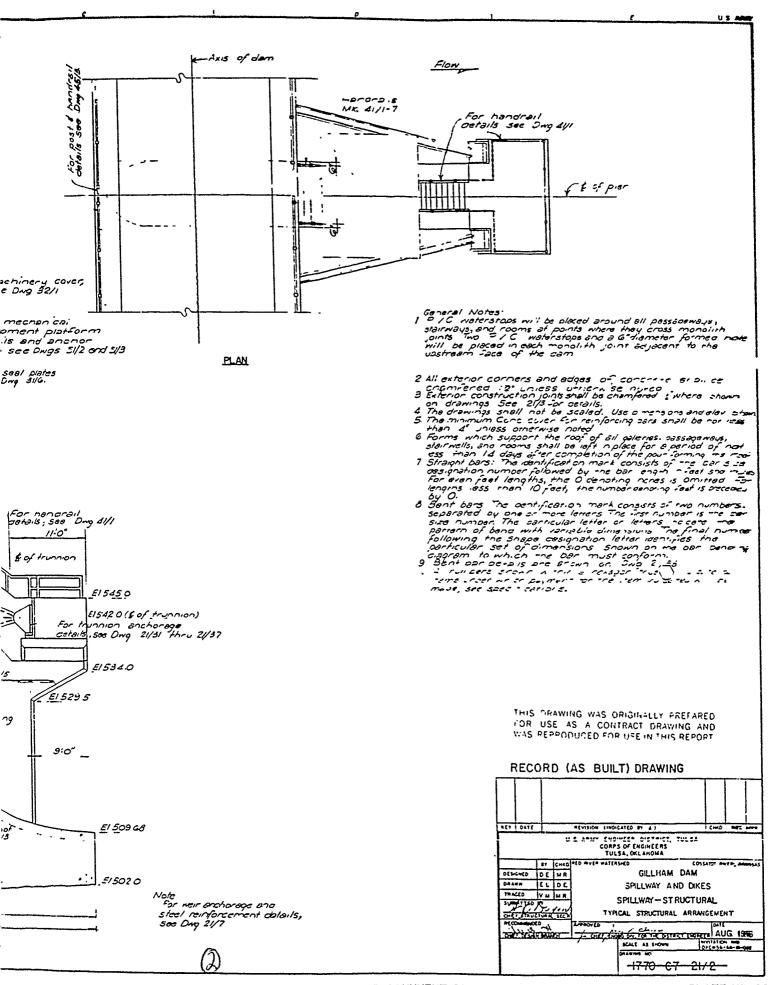


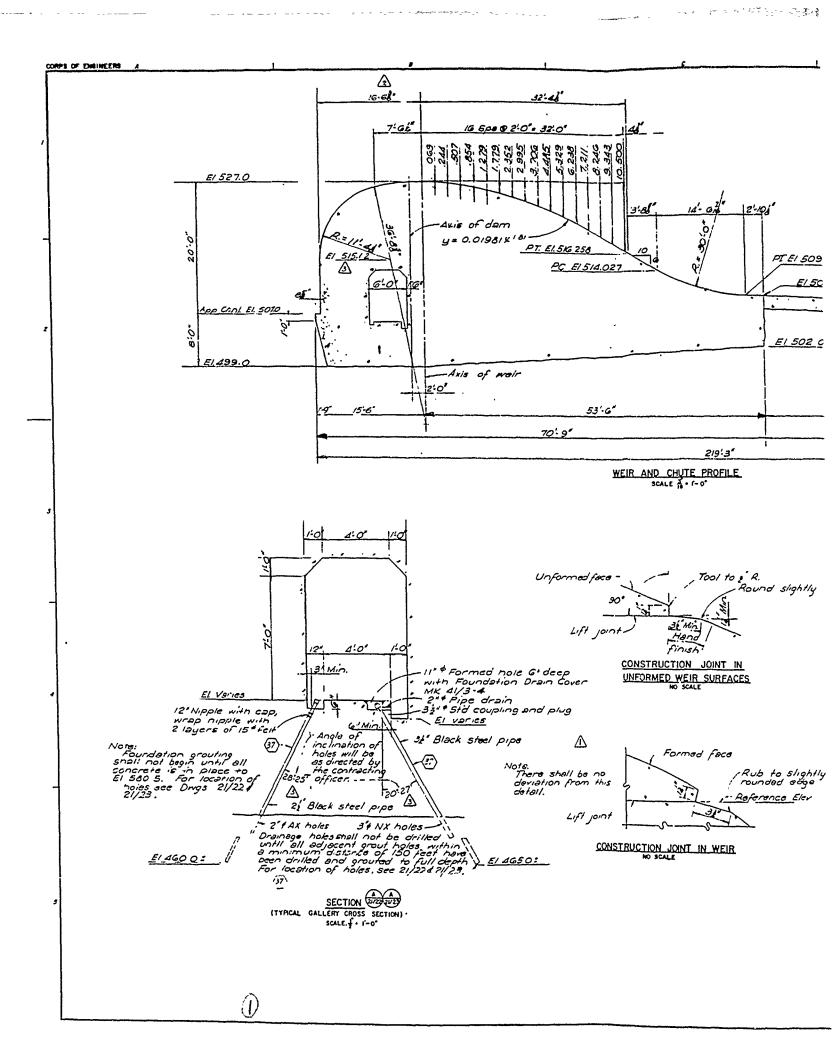


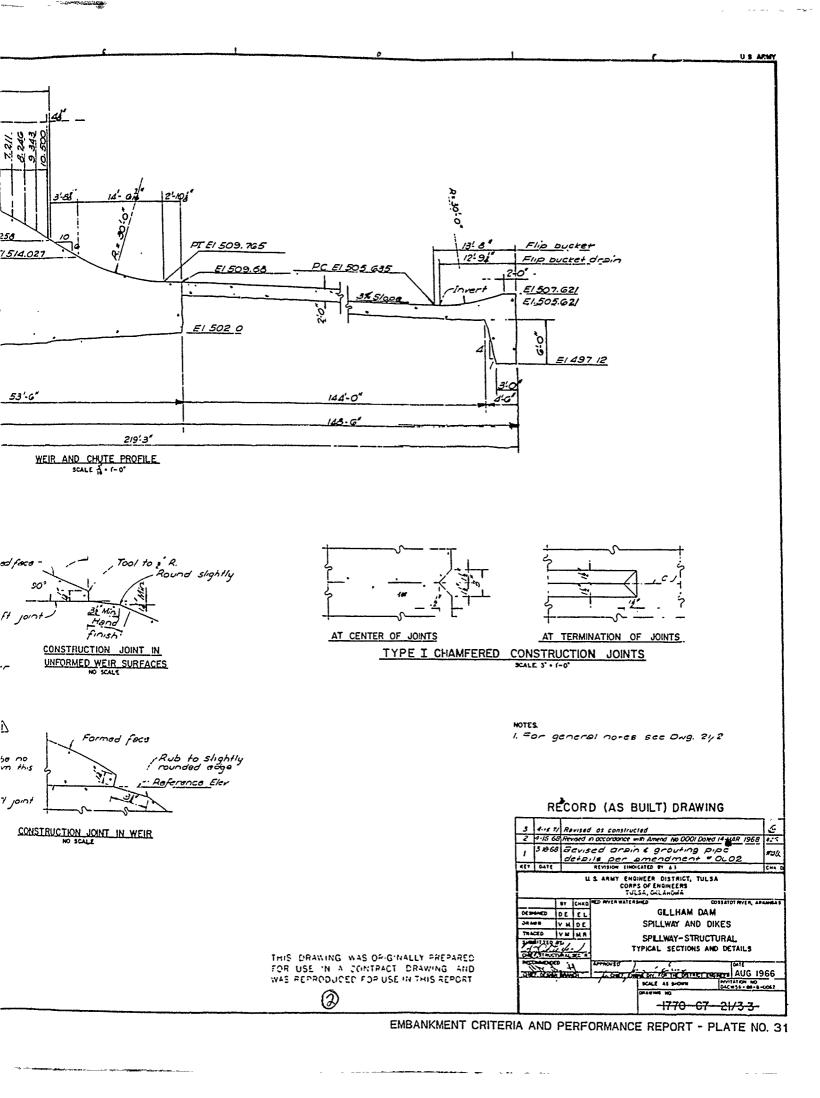
MARKET SHAPE

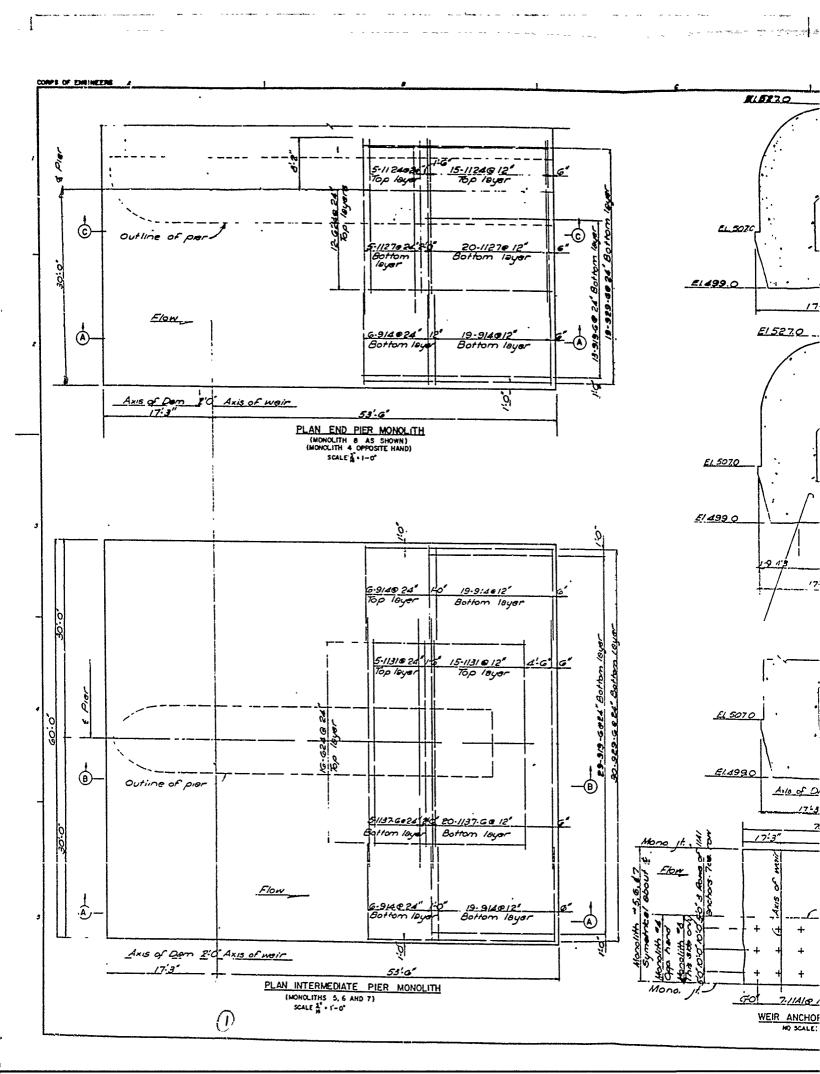
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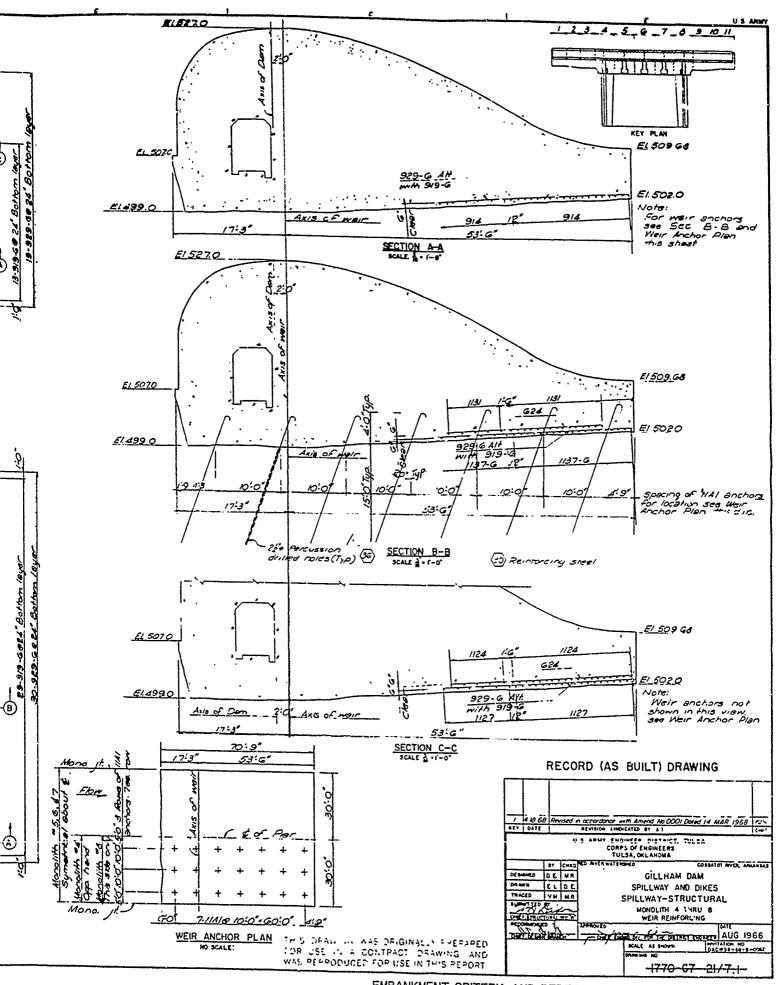


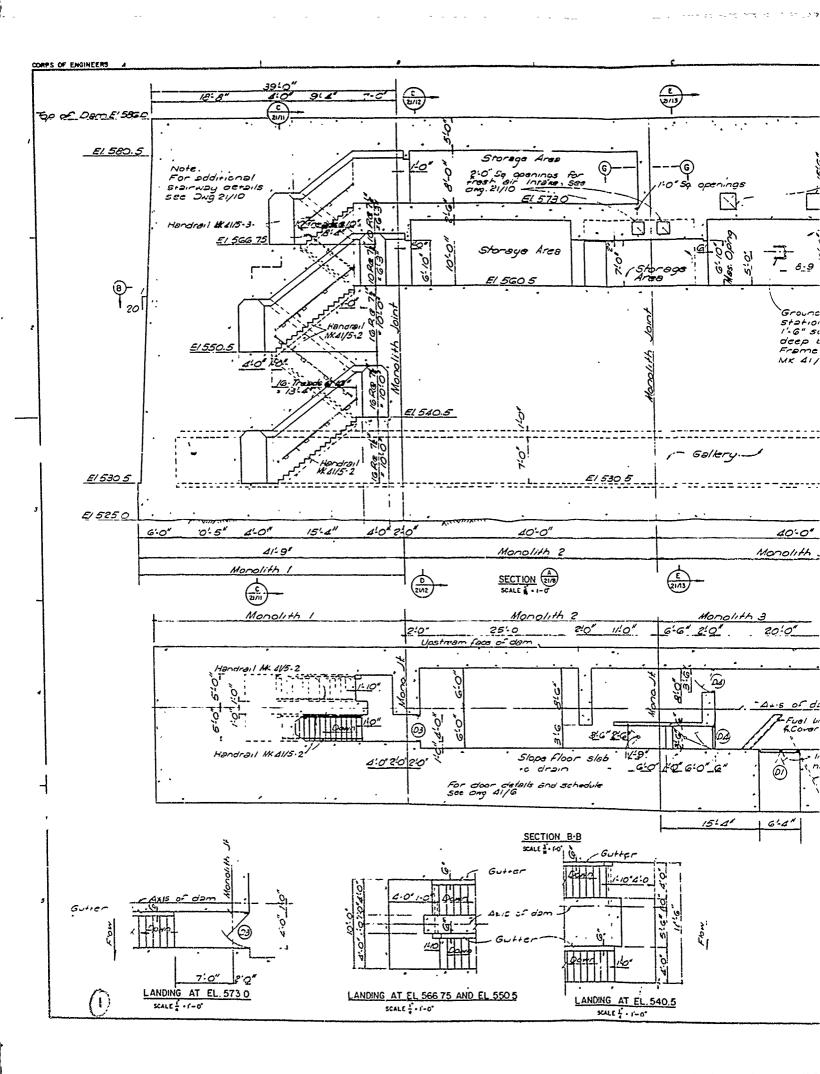


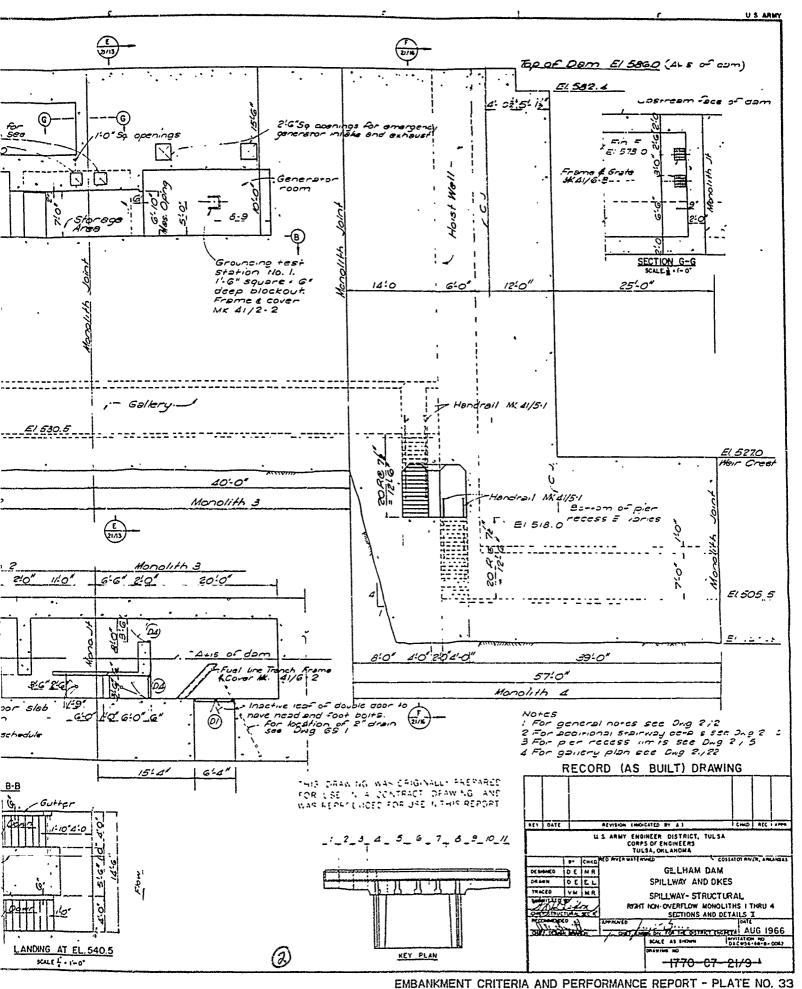


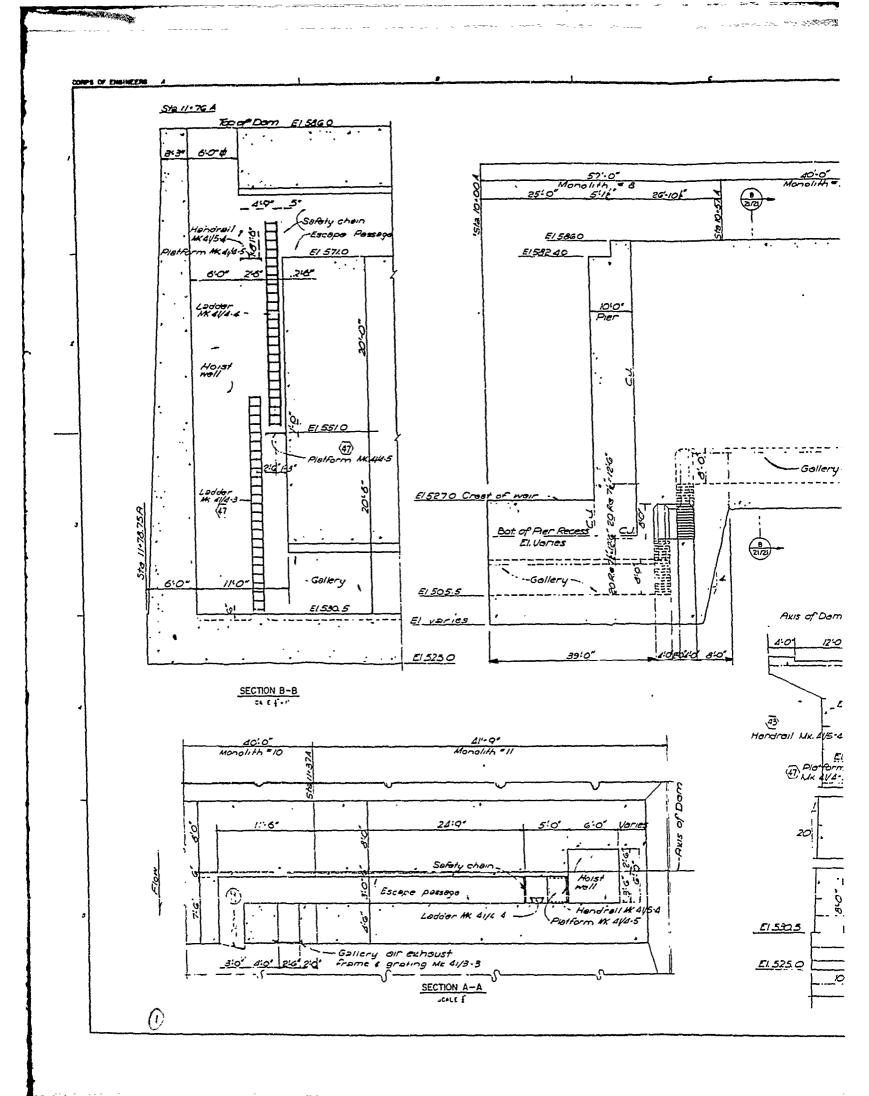


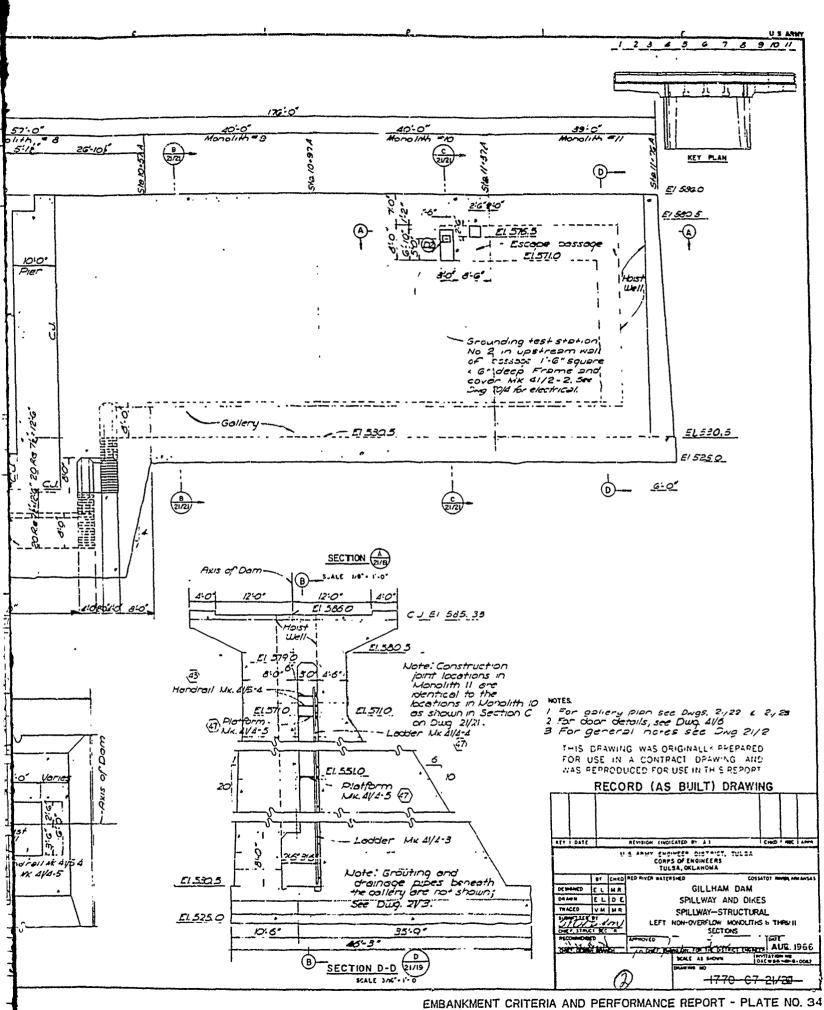


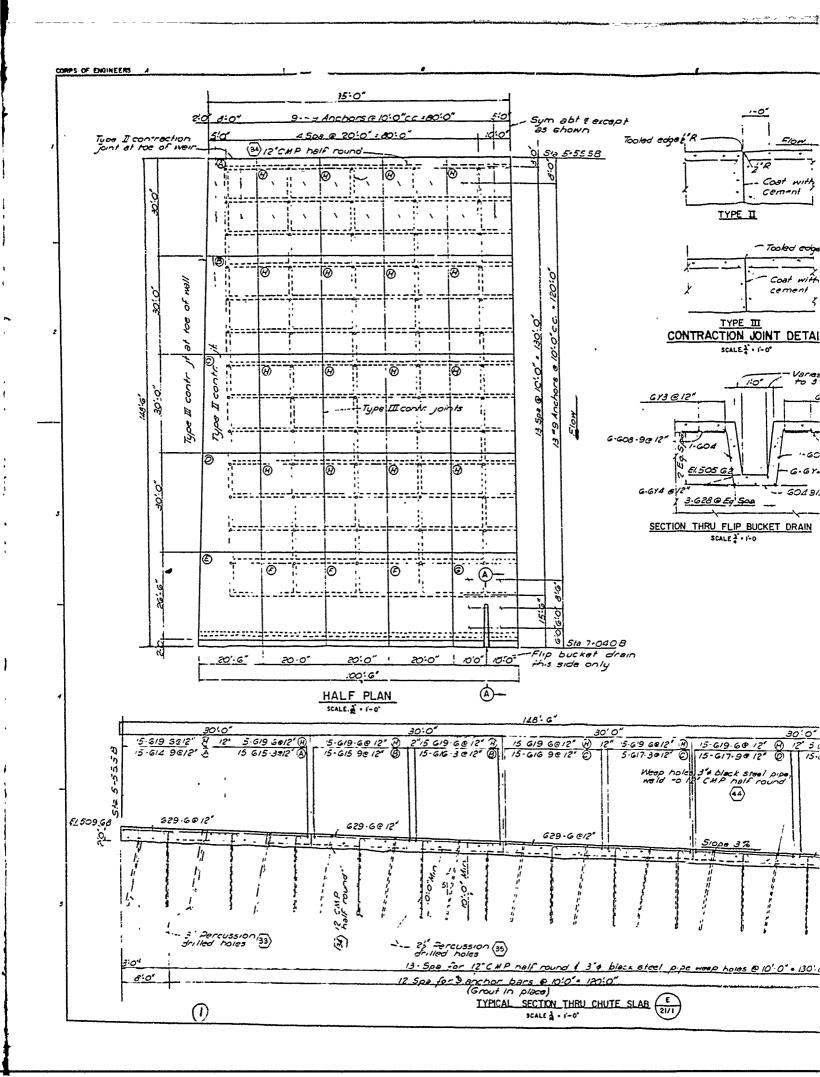




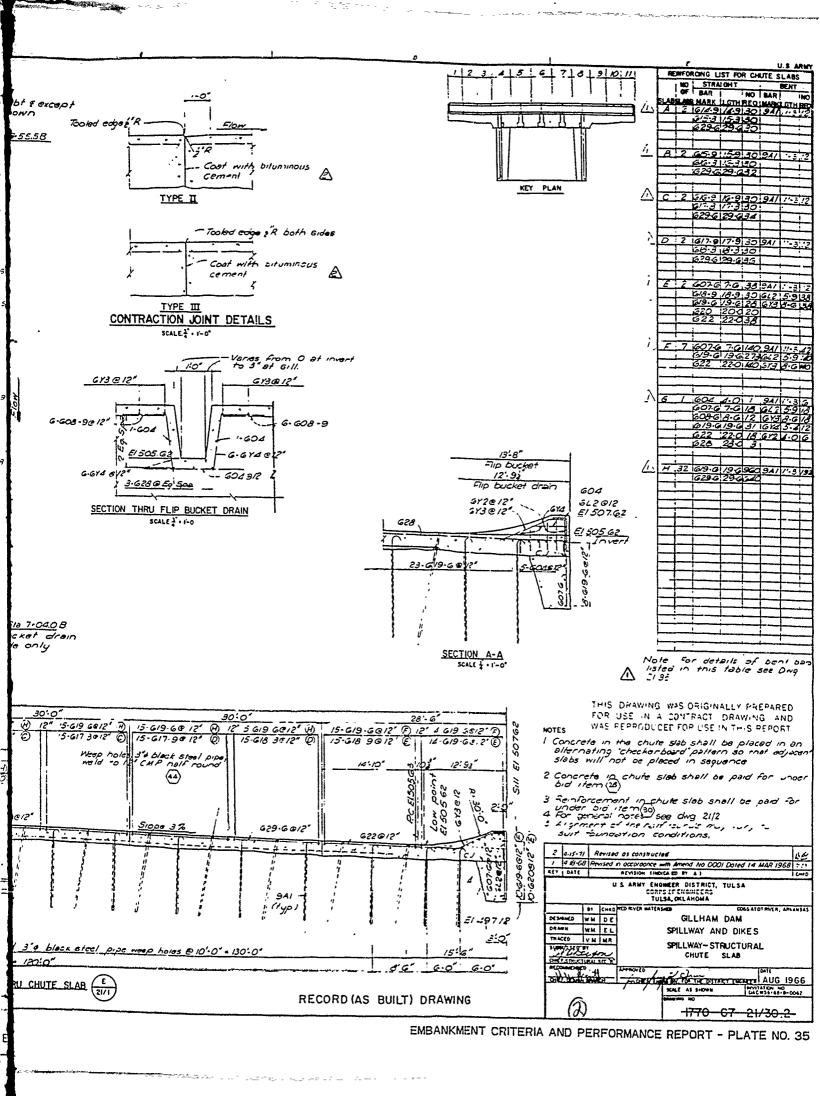


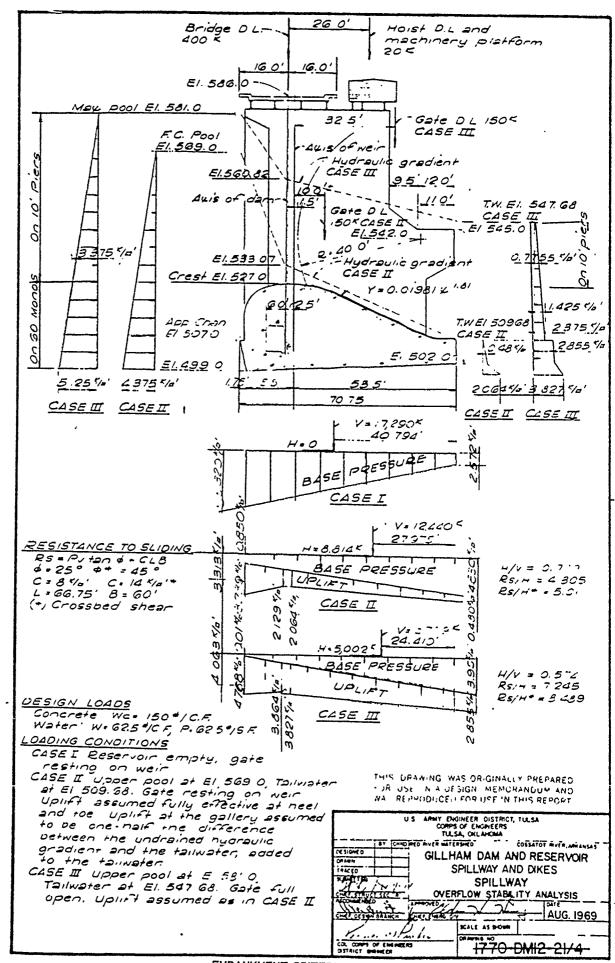




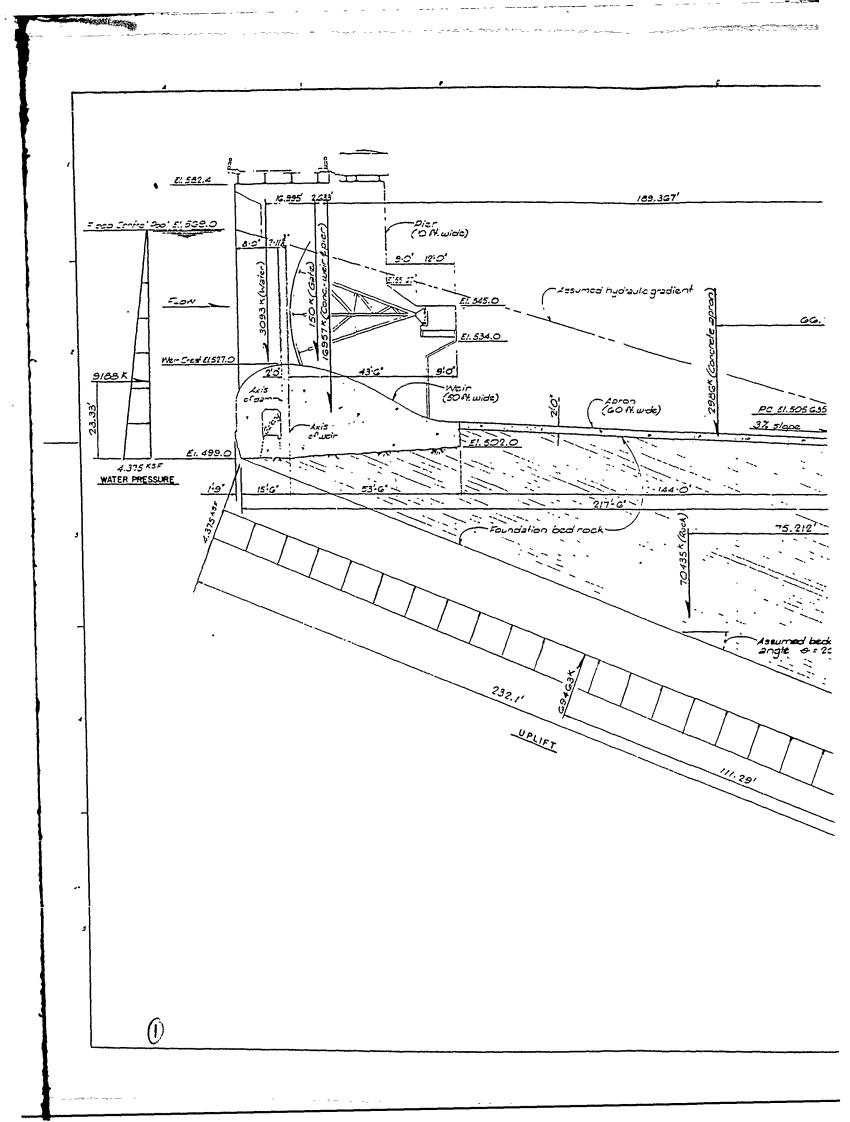


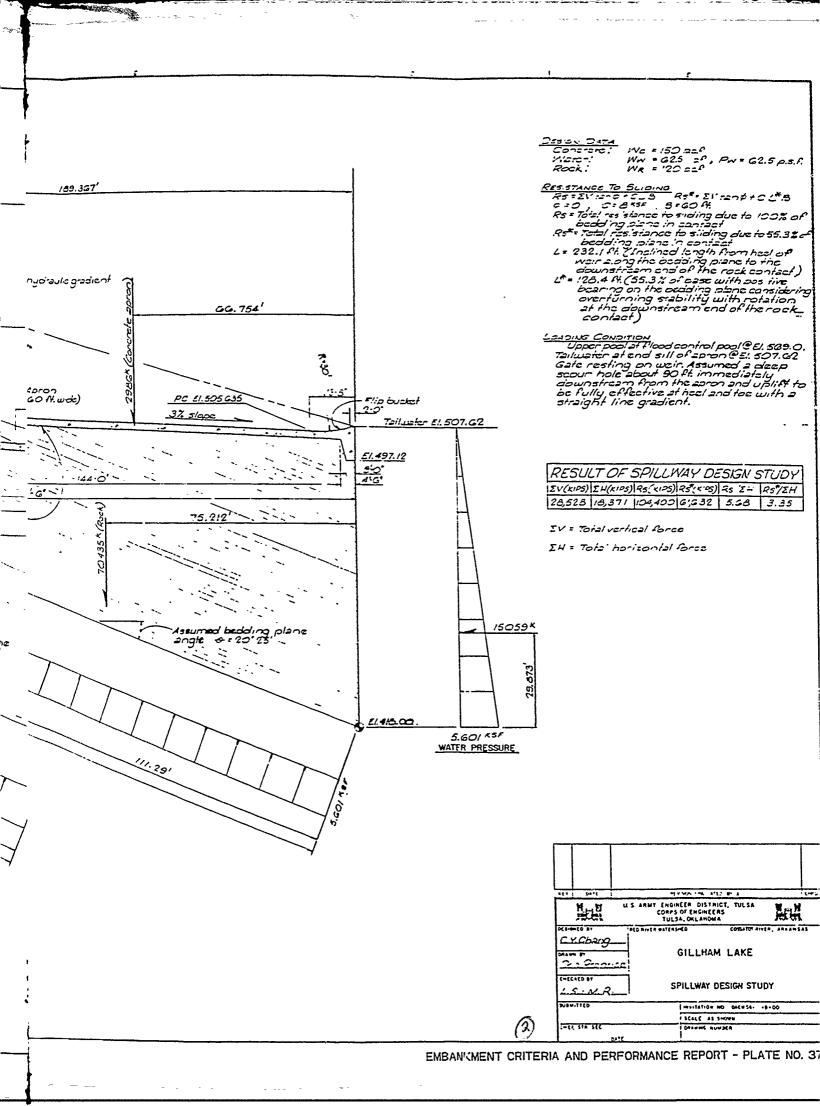
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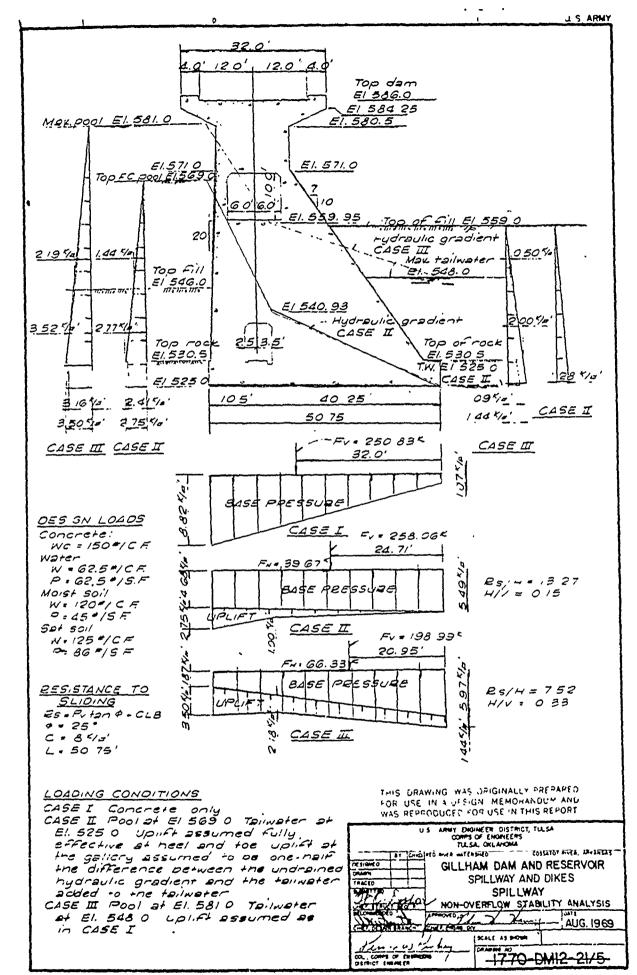




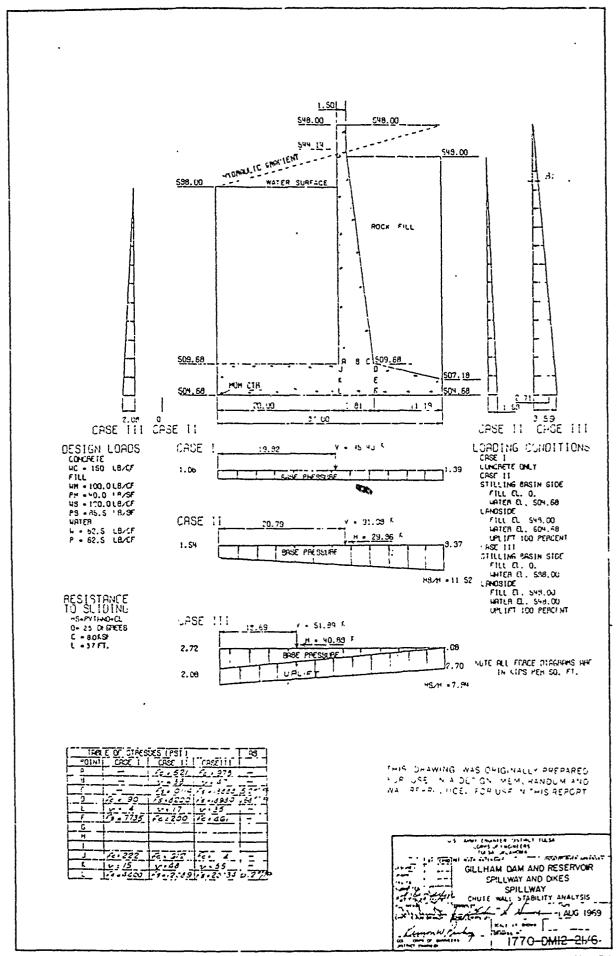
EMBANKMENT CRITERIA AND PERFORMANCE REPORT - PLATE NO. 36

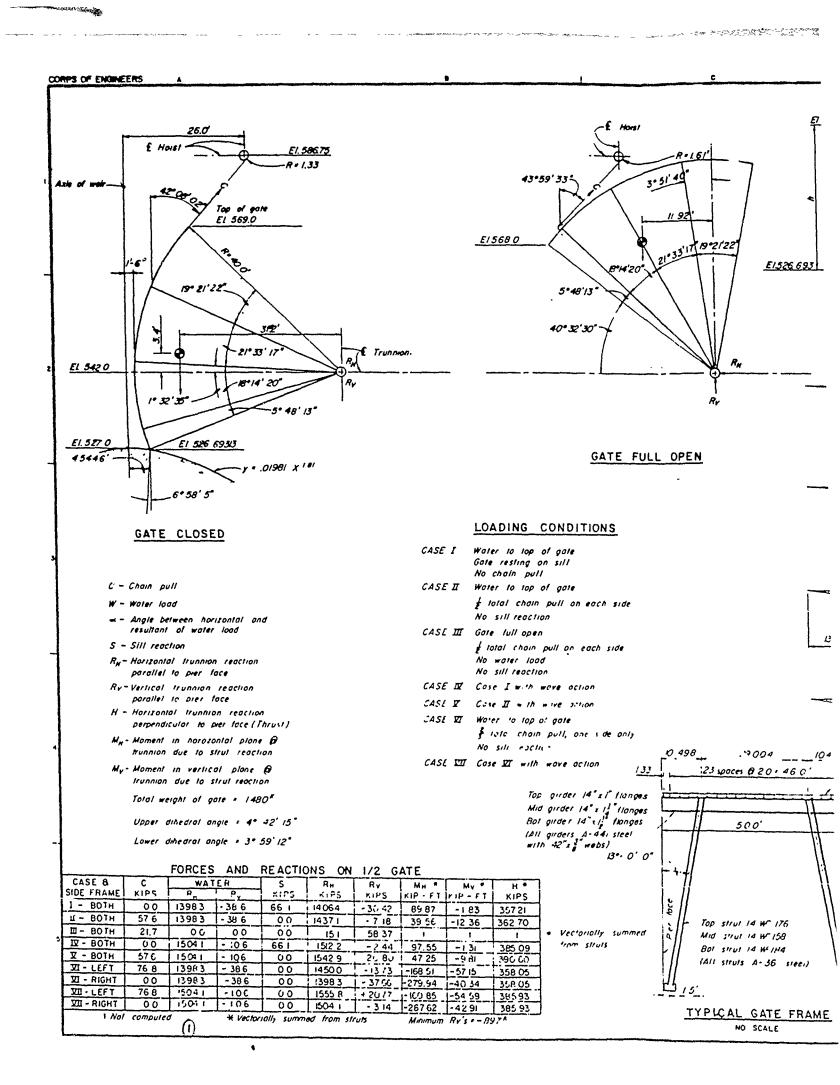


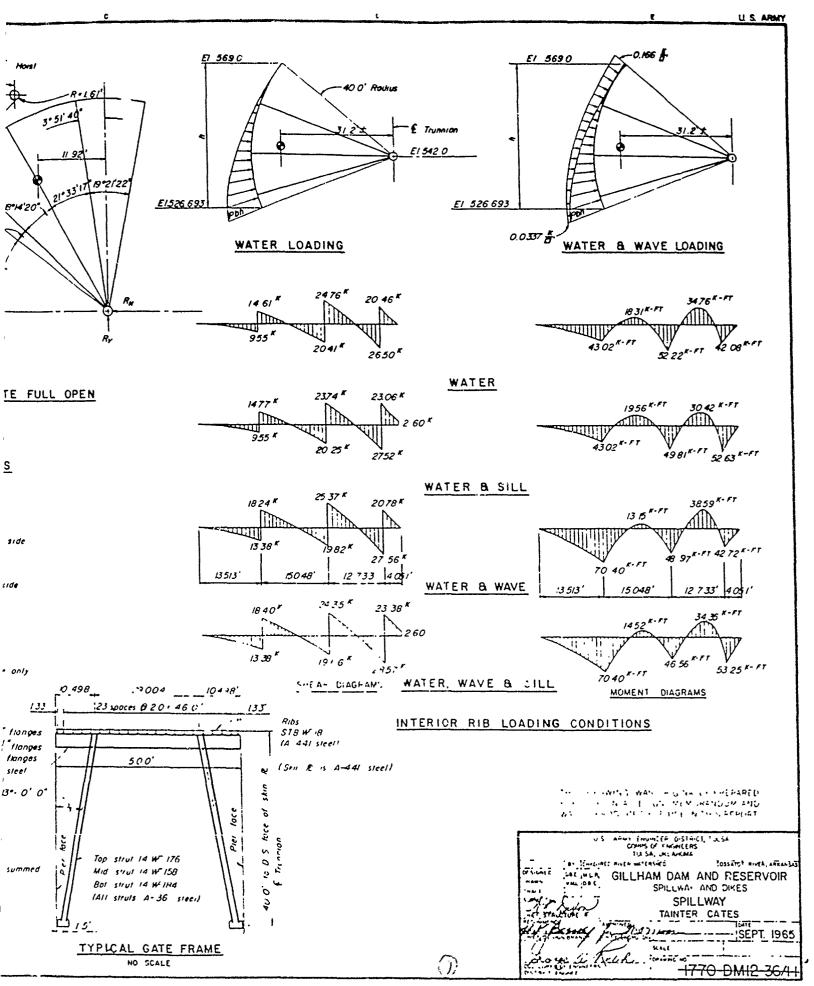




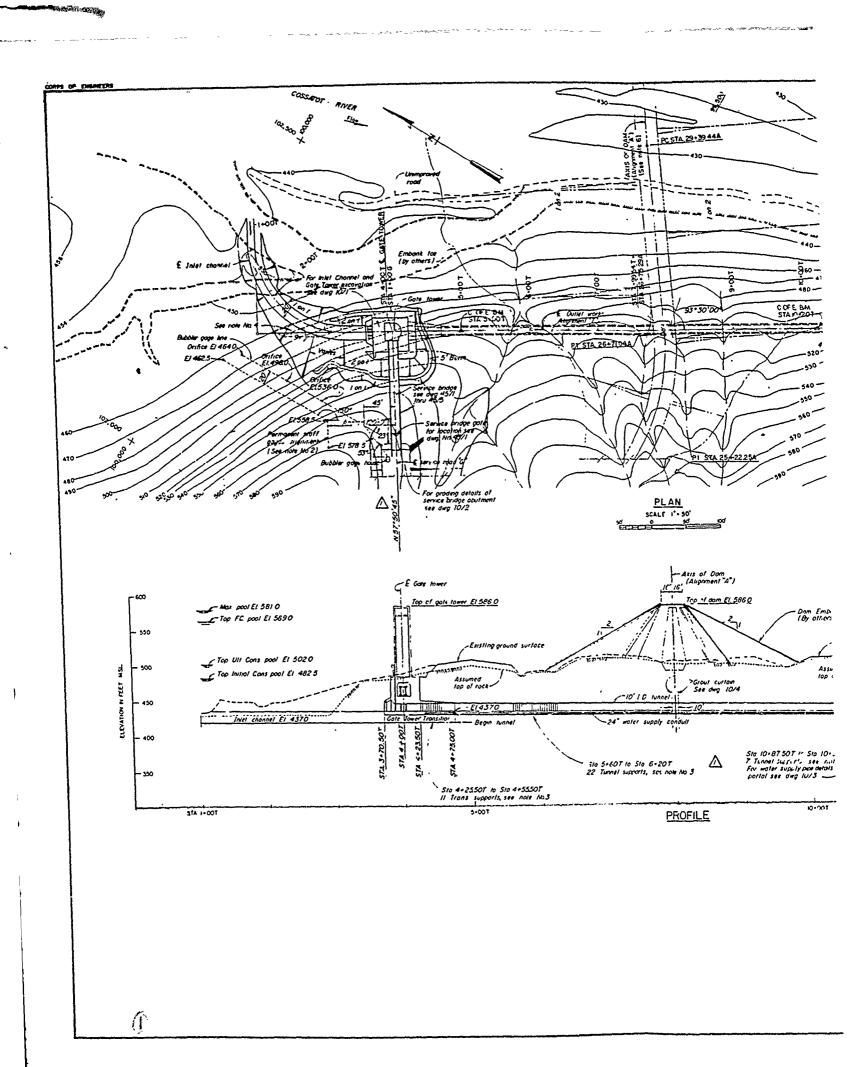
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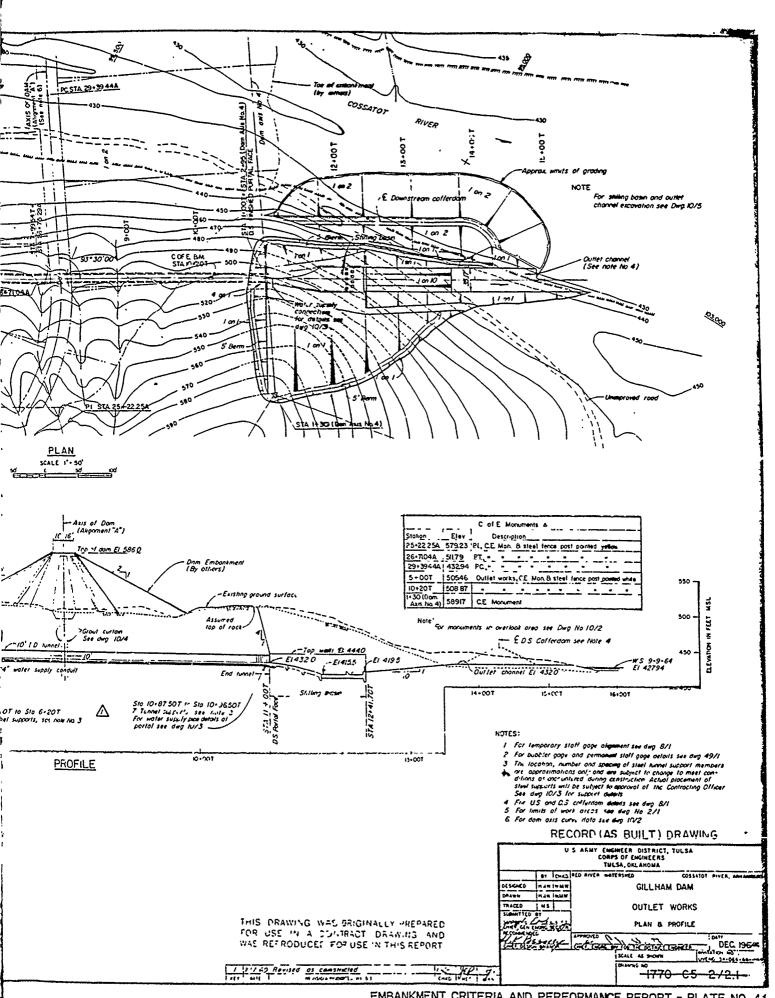






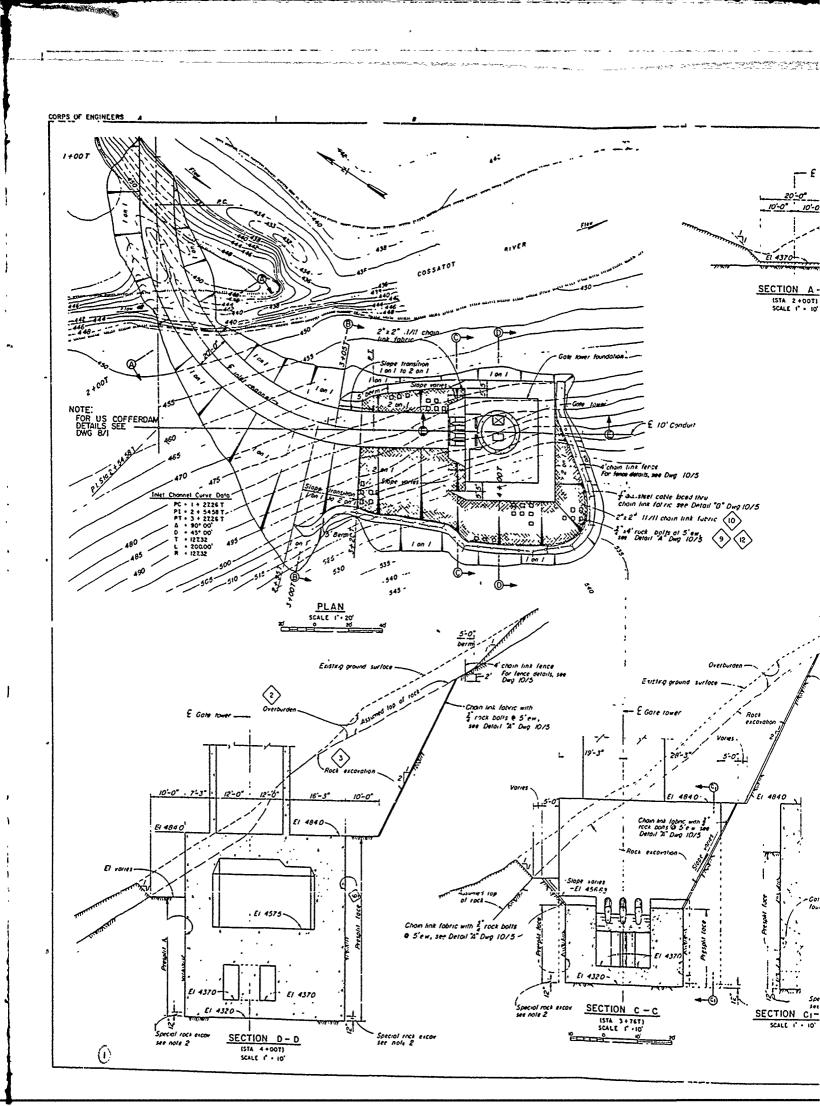
EMBANKMENT CRITERIA AND PERFORMANCE REPORT - PLATE NO. 40

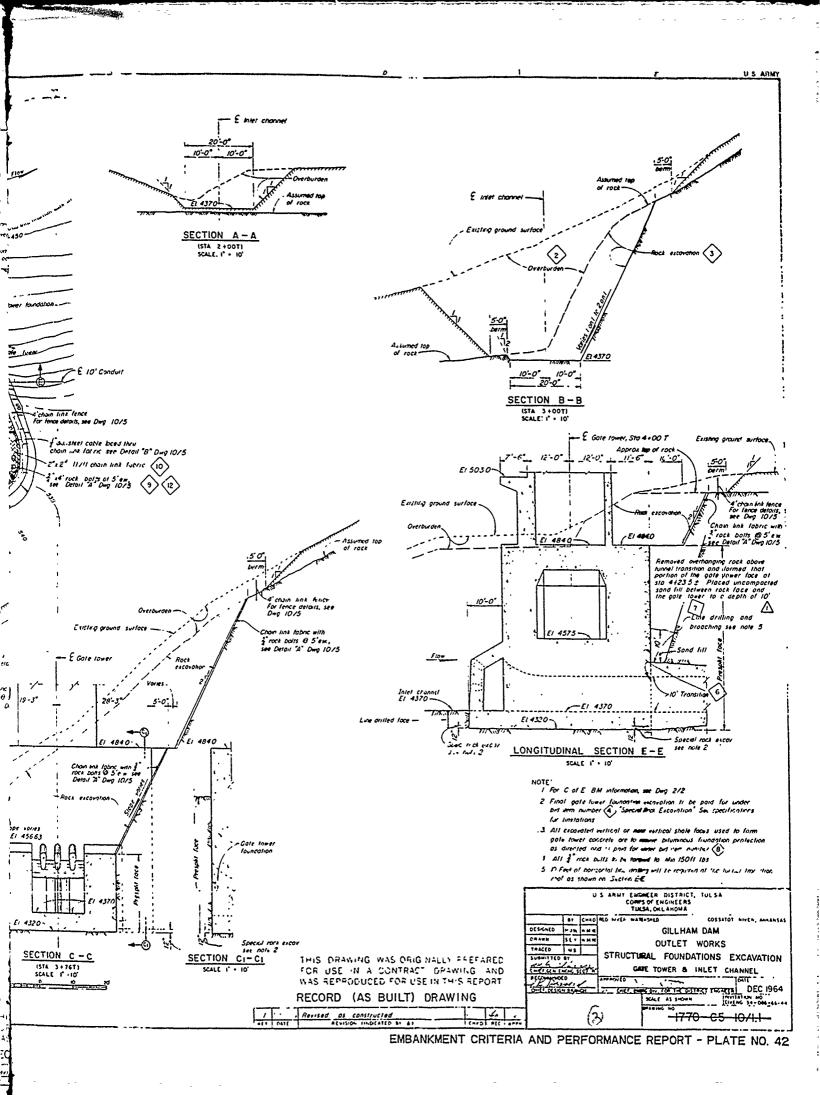


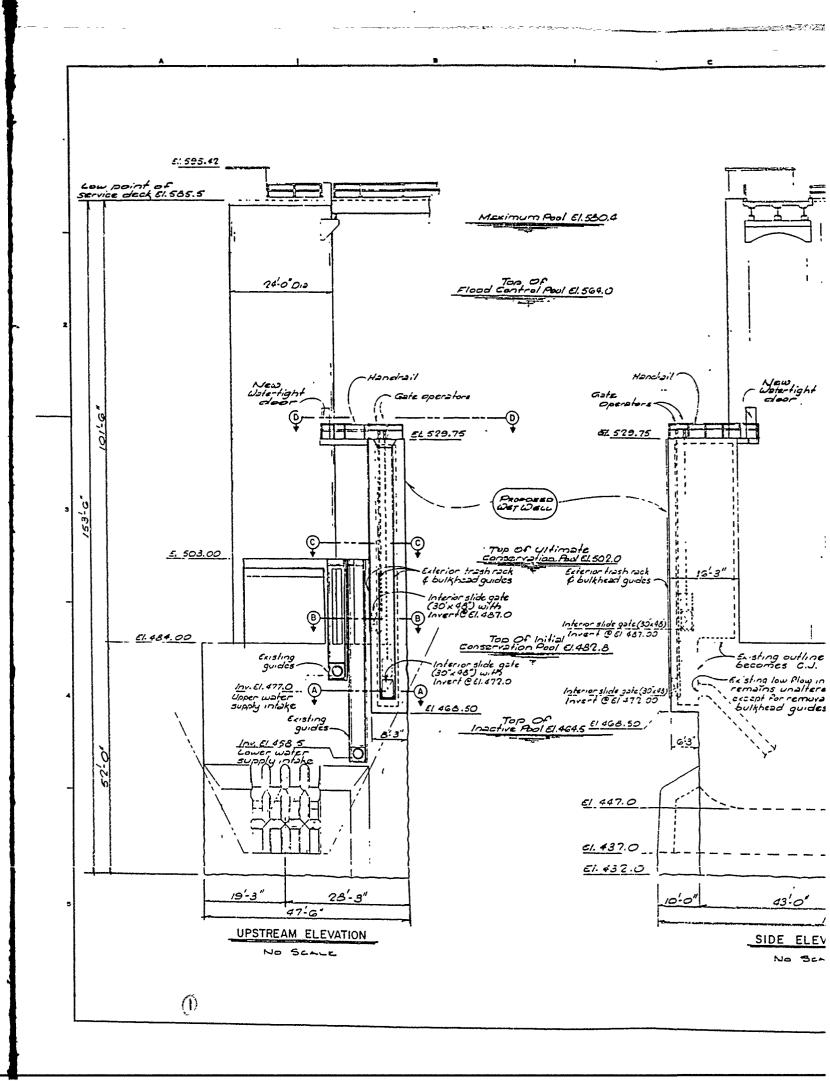


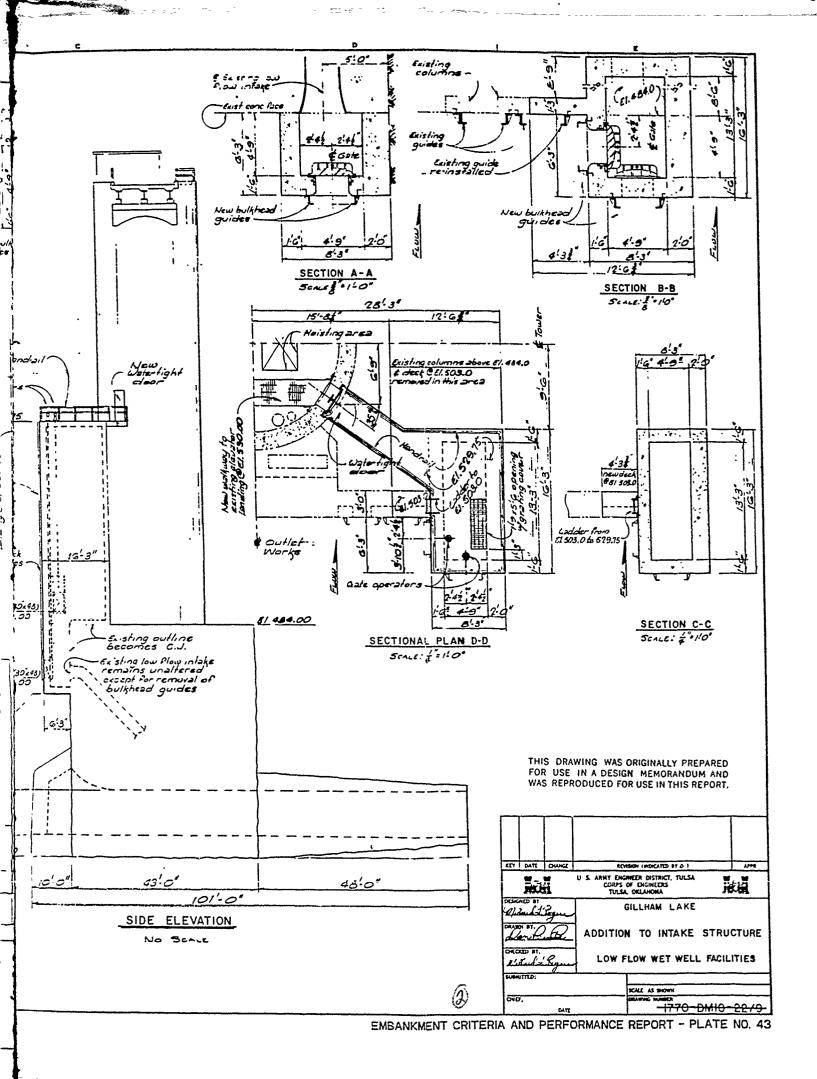
STATE STATES

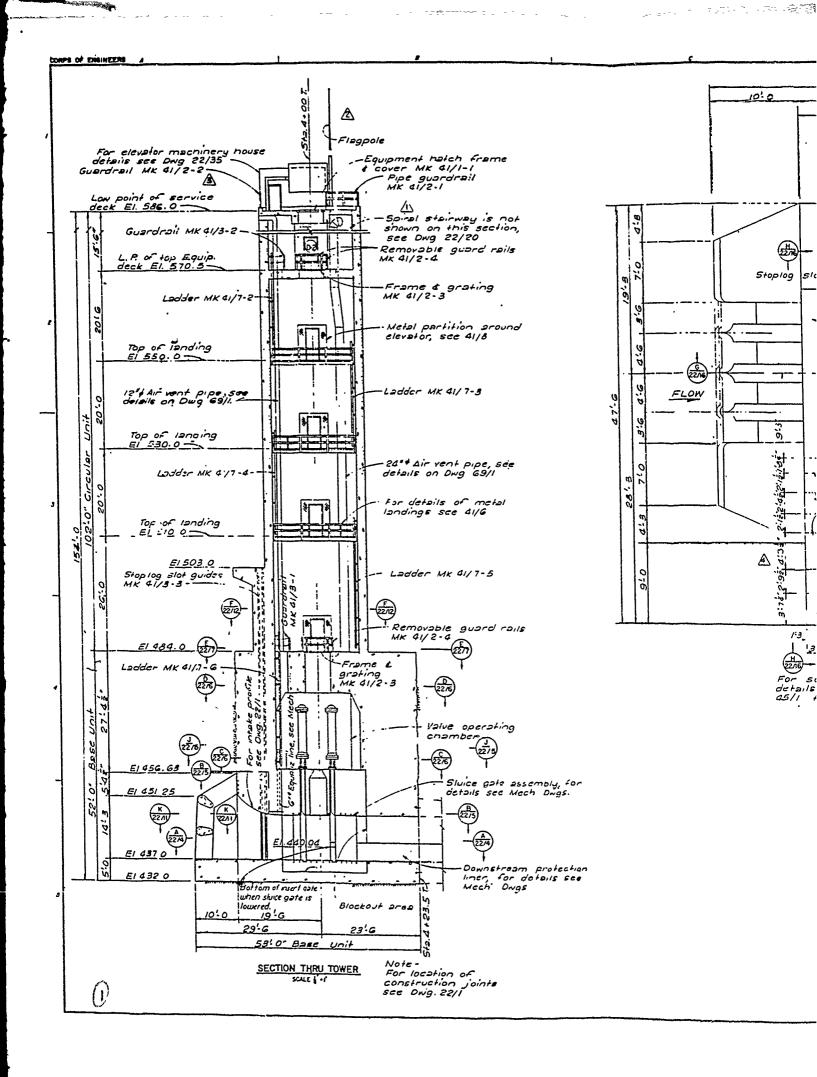
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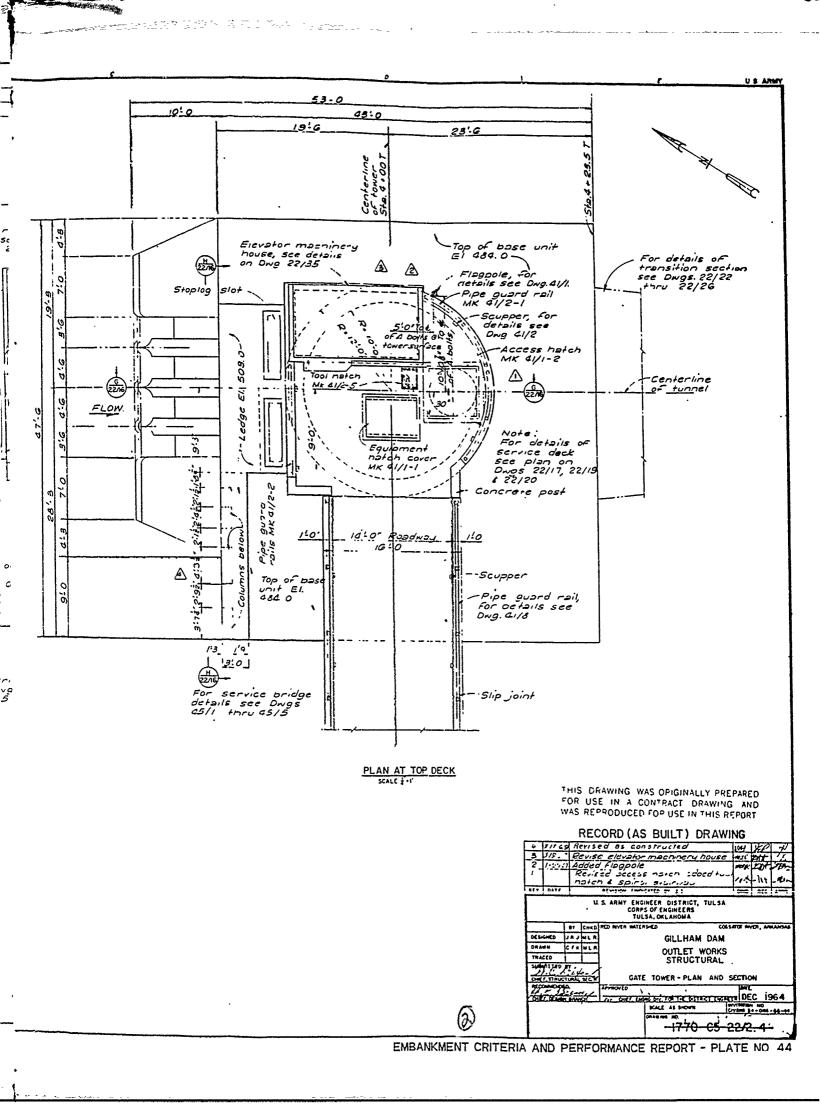


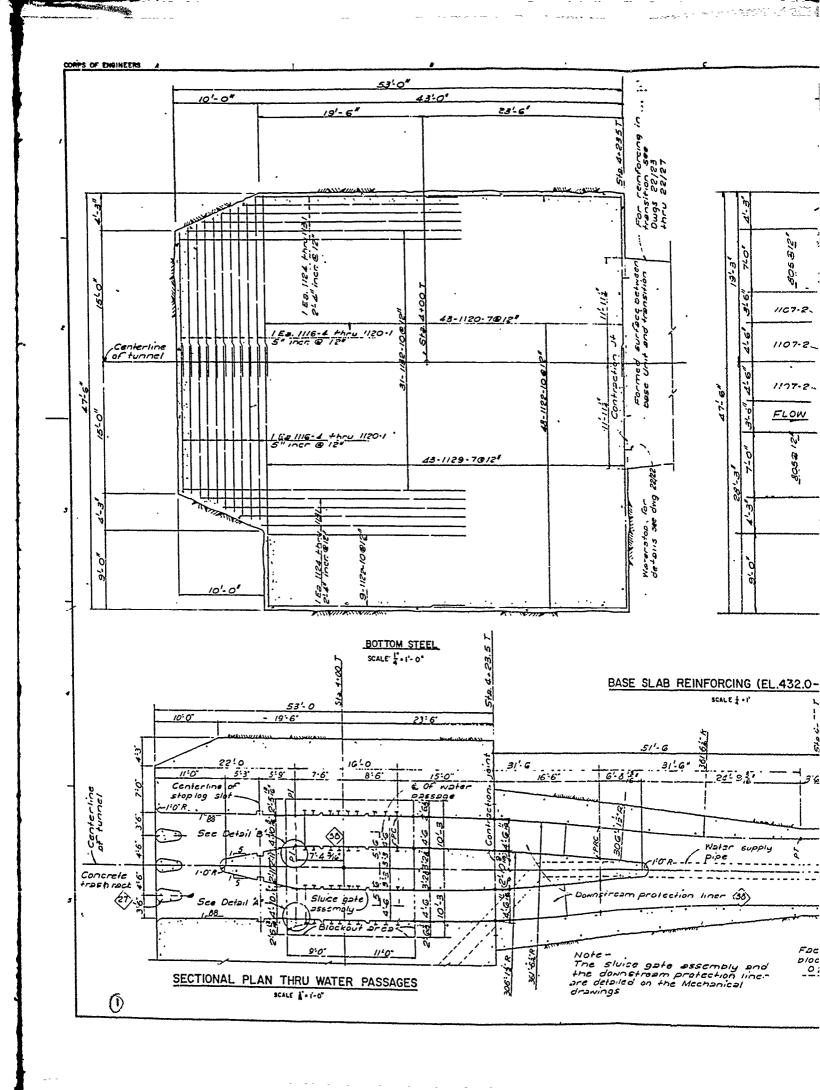


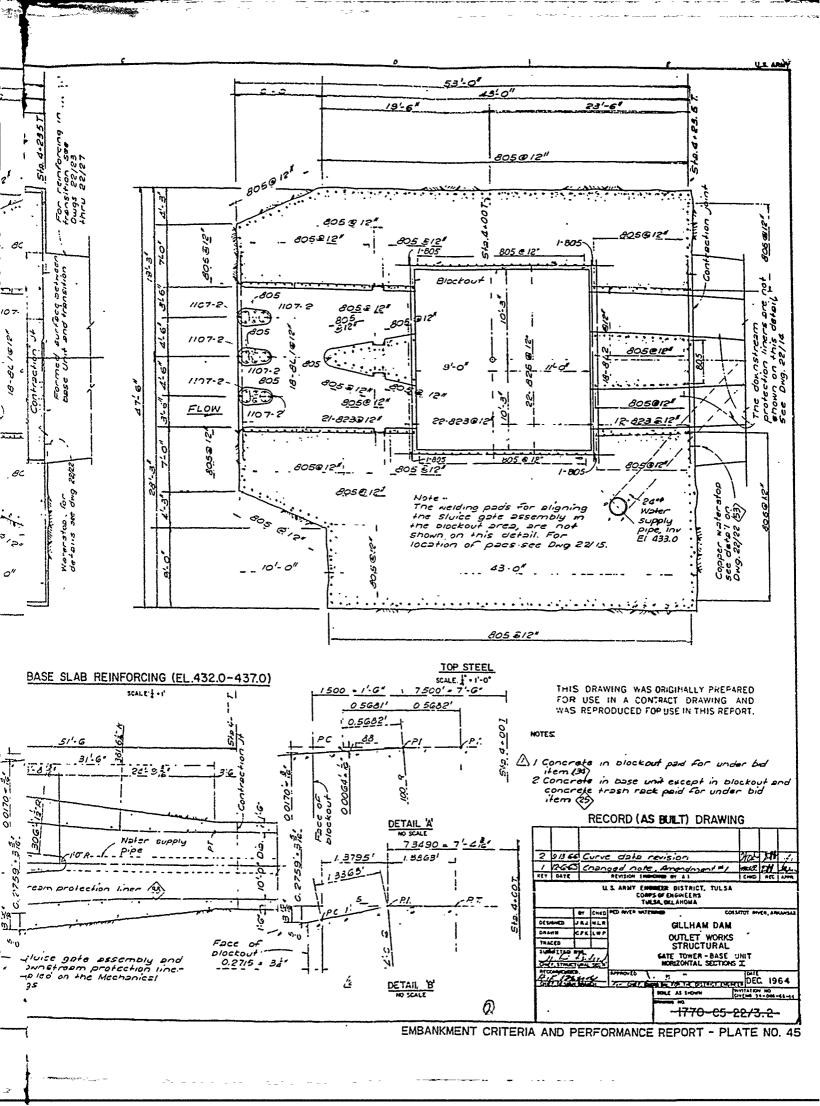


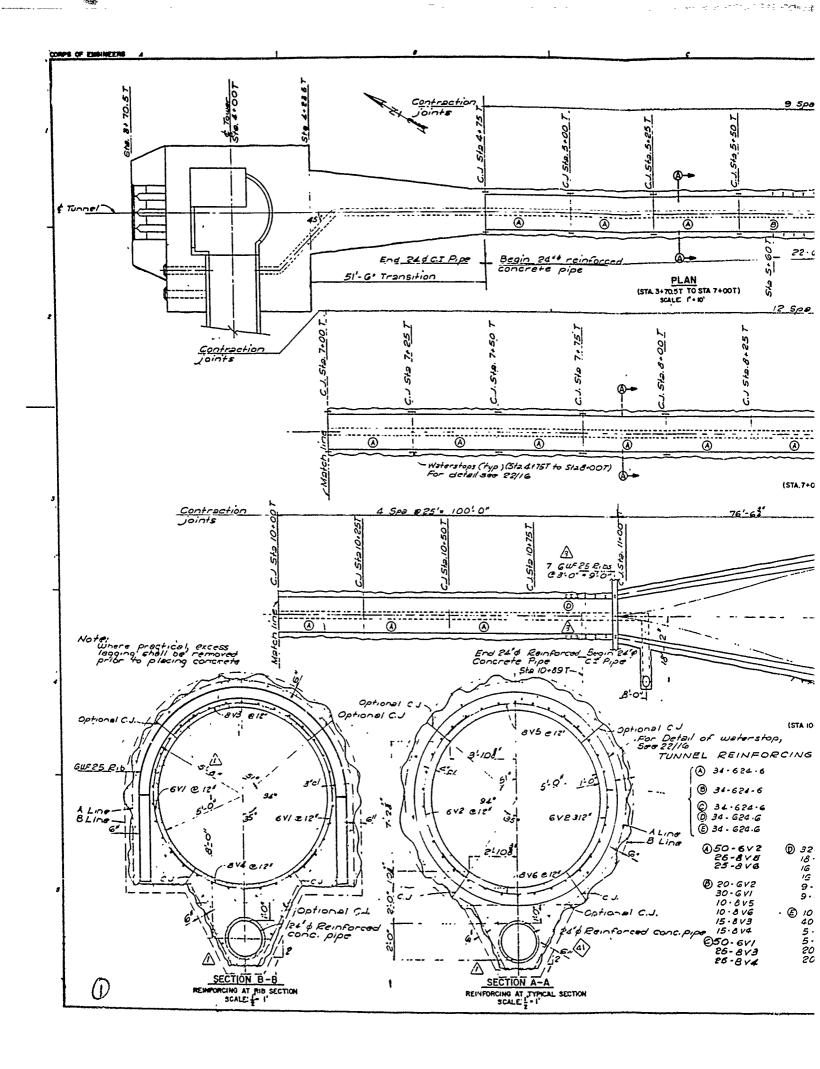


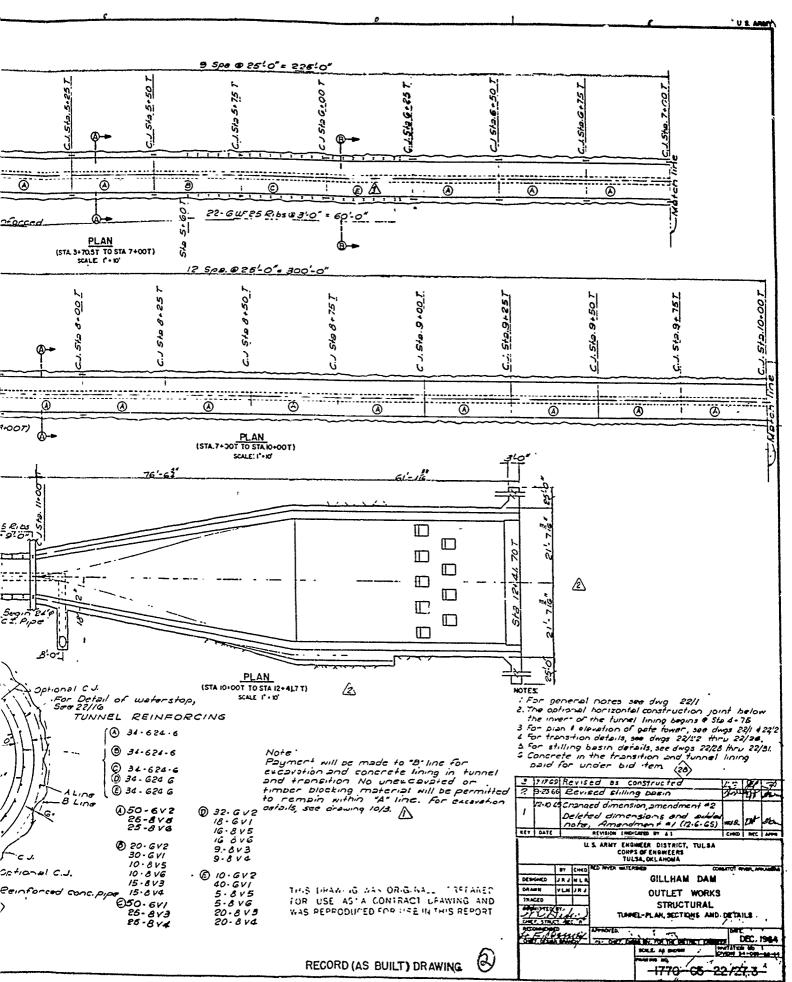


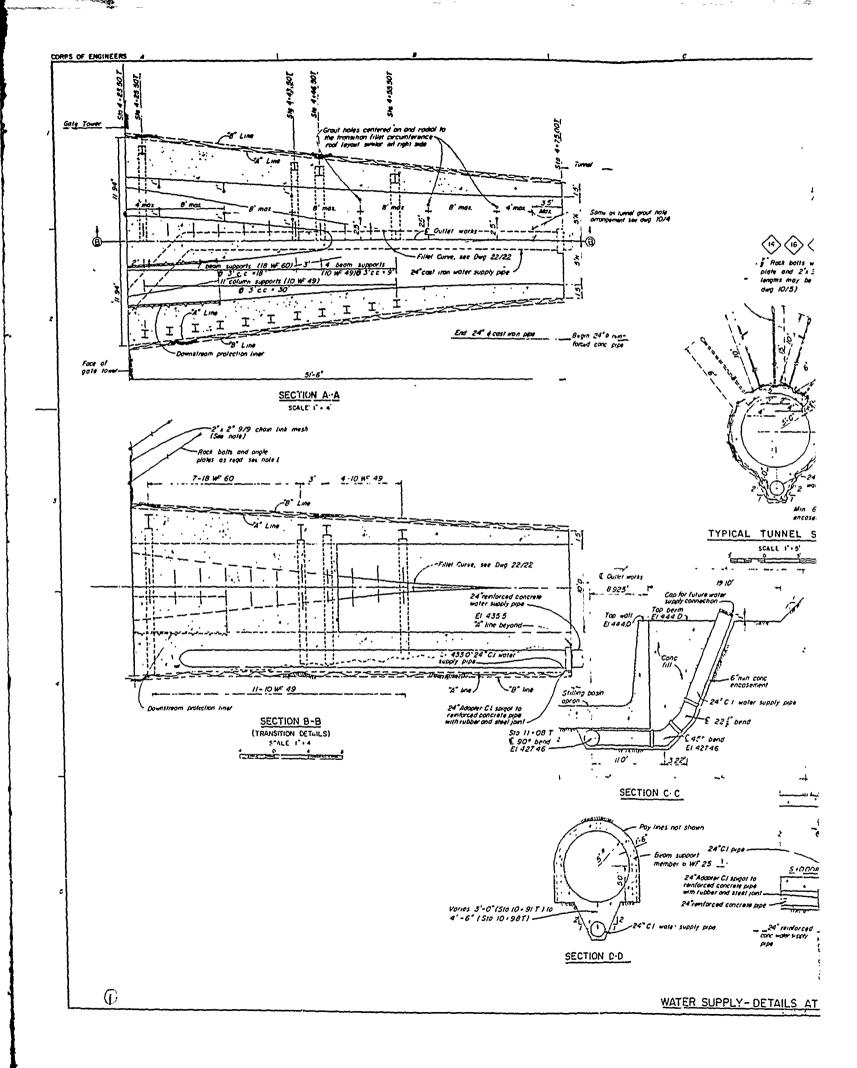


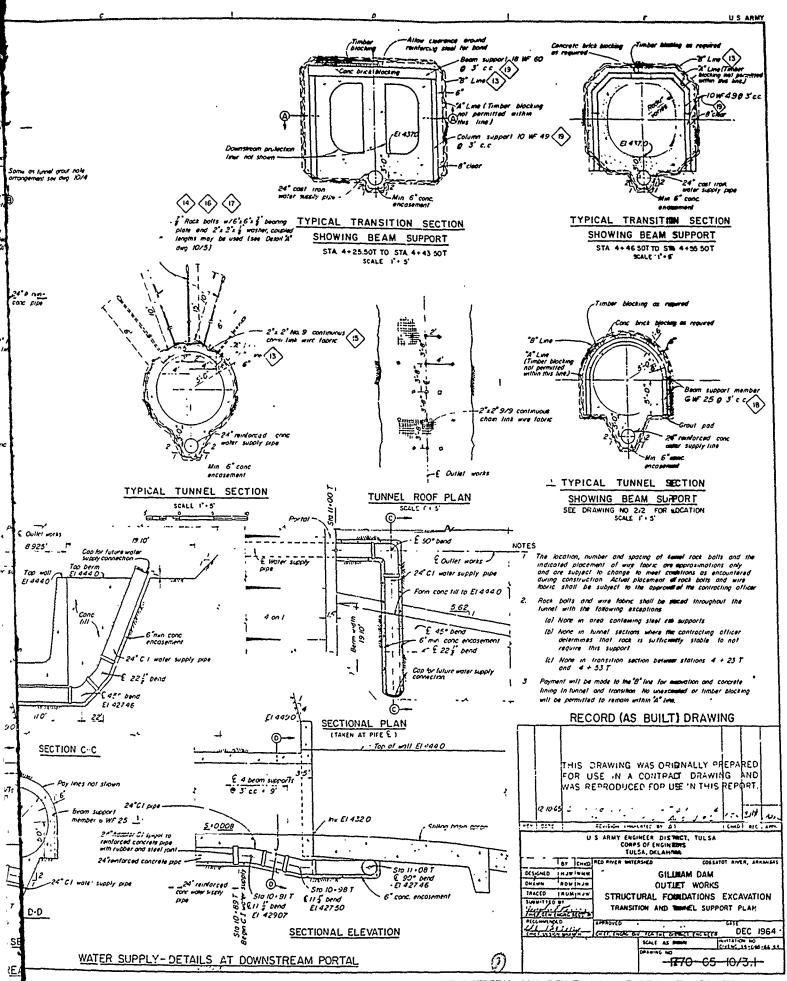








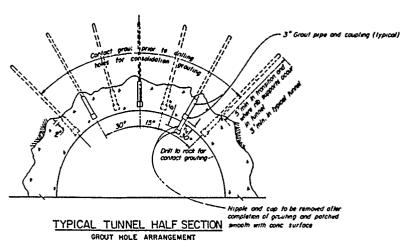




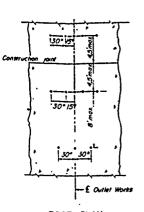
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CORPS OF ENGINEERS

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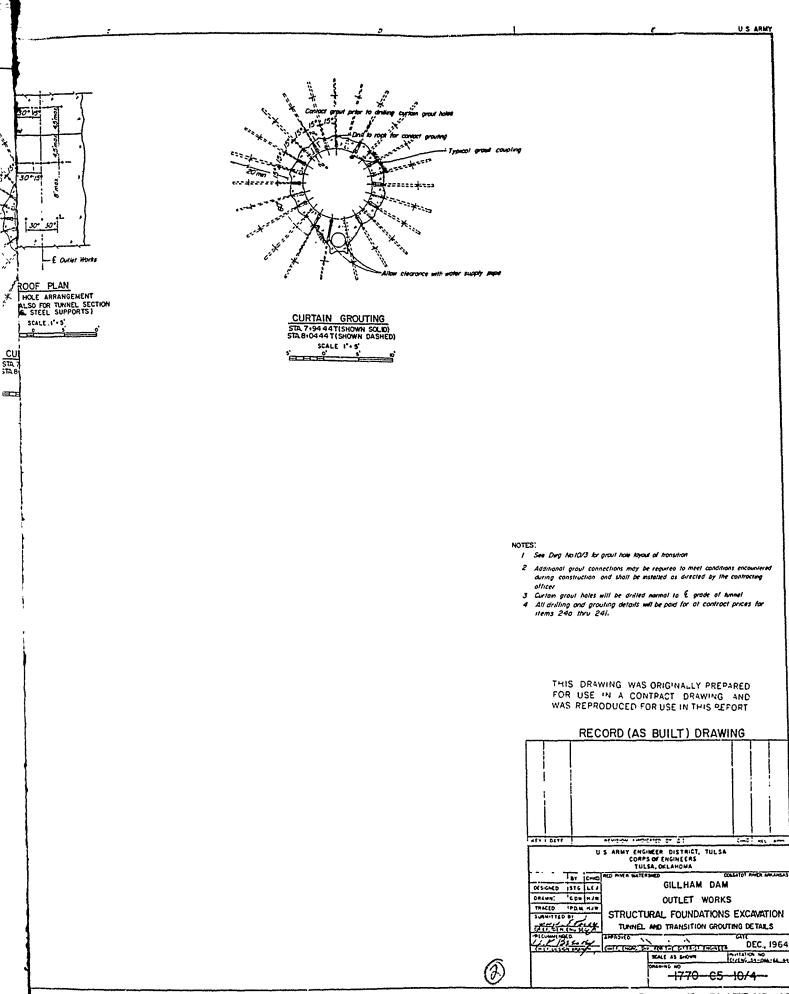
GROUT HOLE ARRANGEMENT
SCALE 1 2

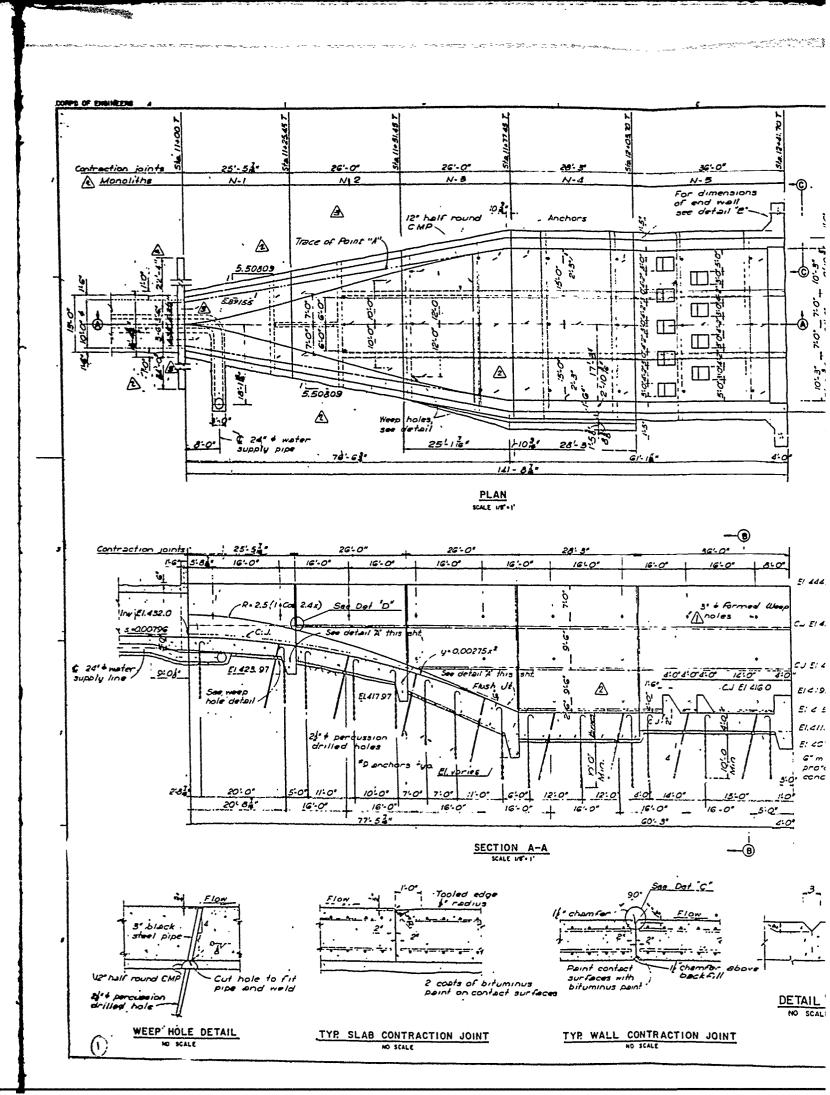


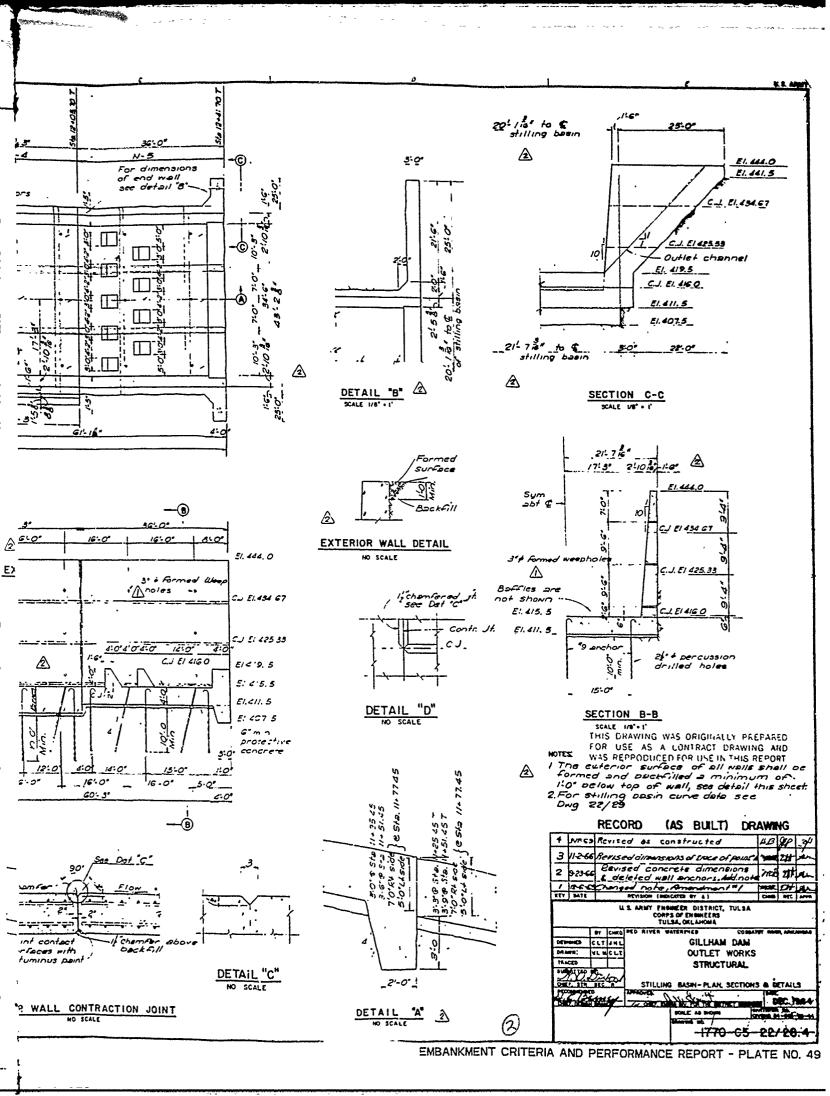
And the second s

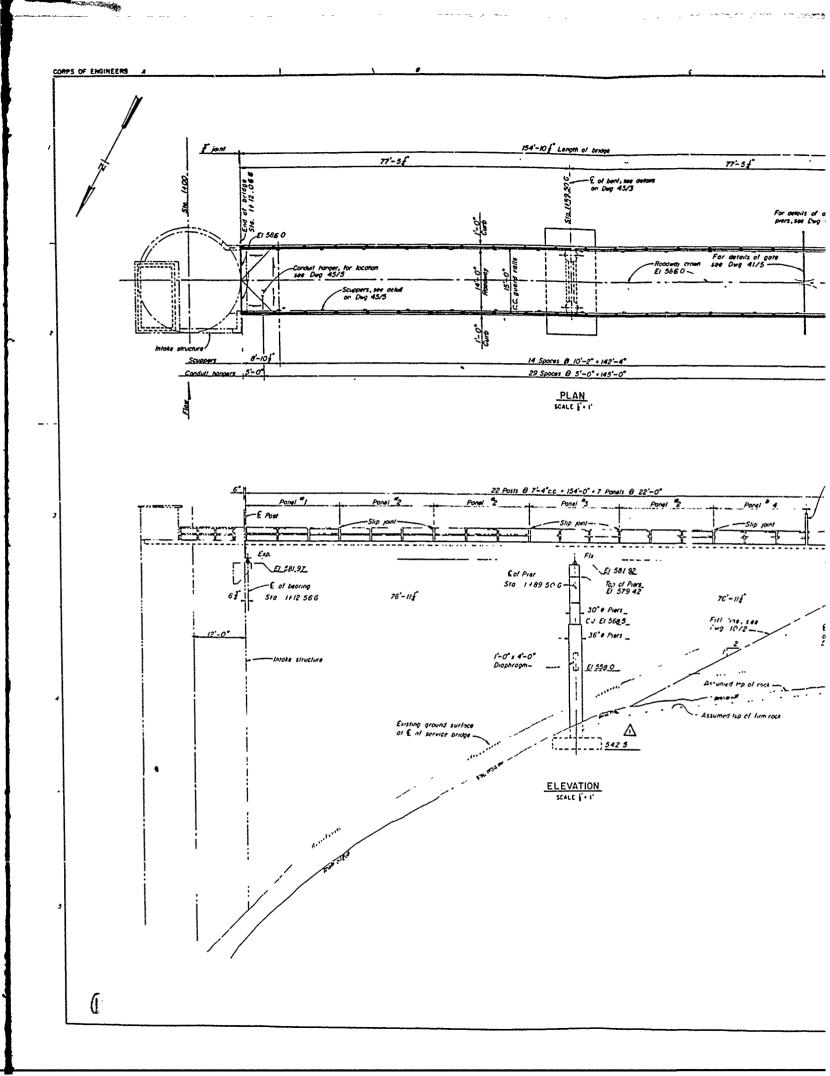
ROOF PLAN
GROUT HOLE ARRANGEMENT
(TYPICAL ALSO FOR TUNNEL SECTION
RECUIRING STEEL SUPPORTS)
SCALE 1. 5

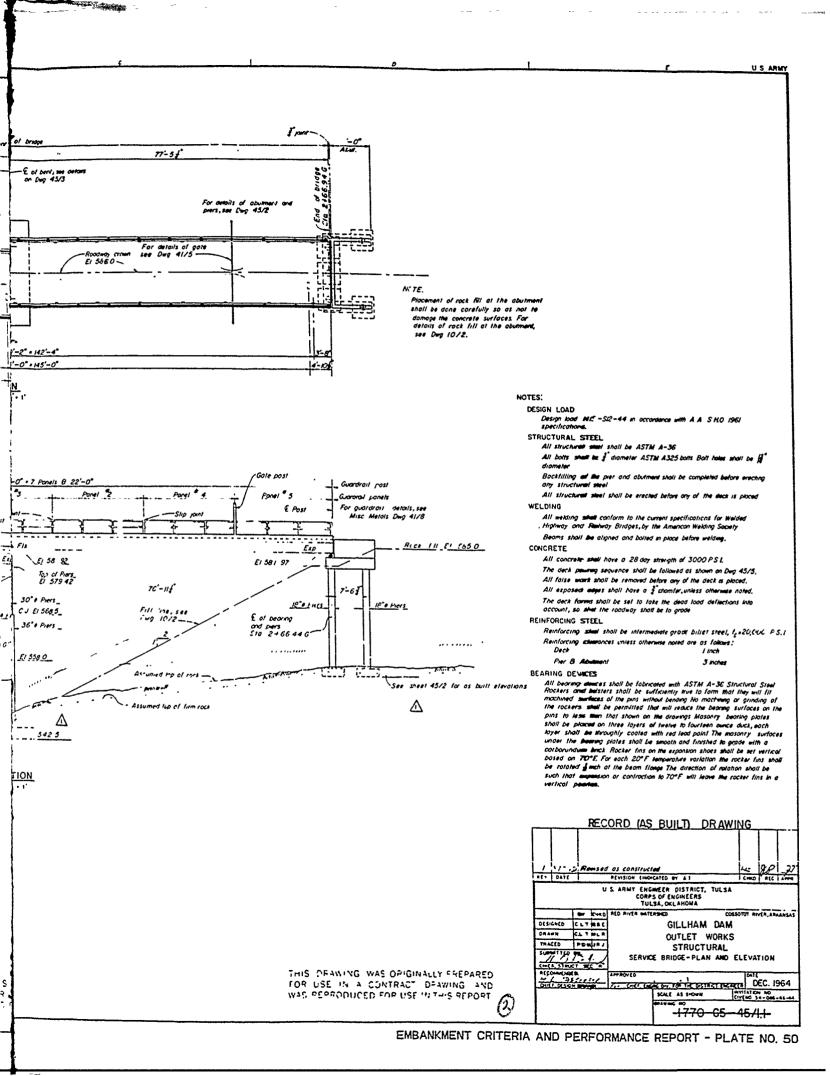
 $(i_{\prime}$

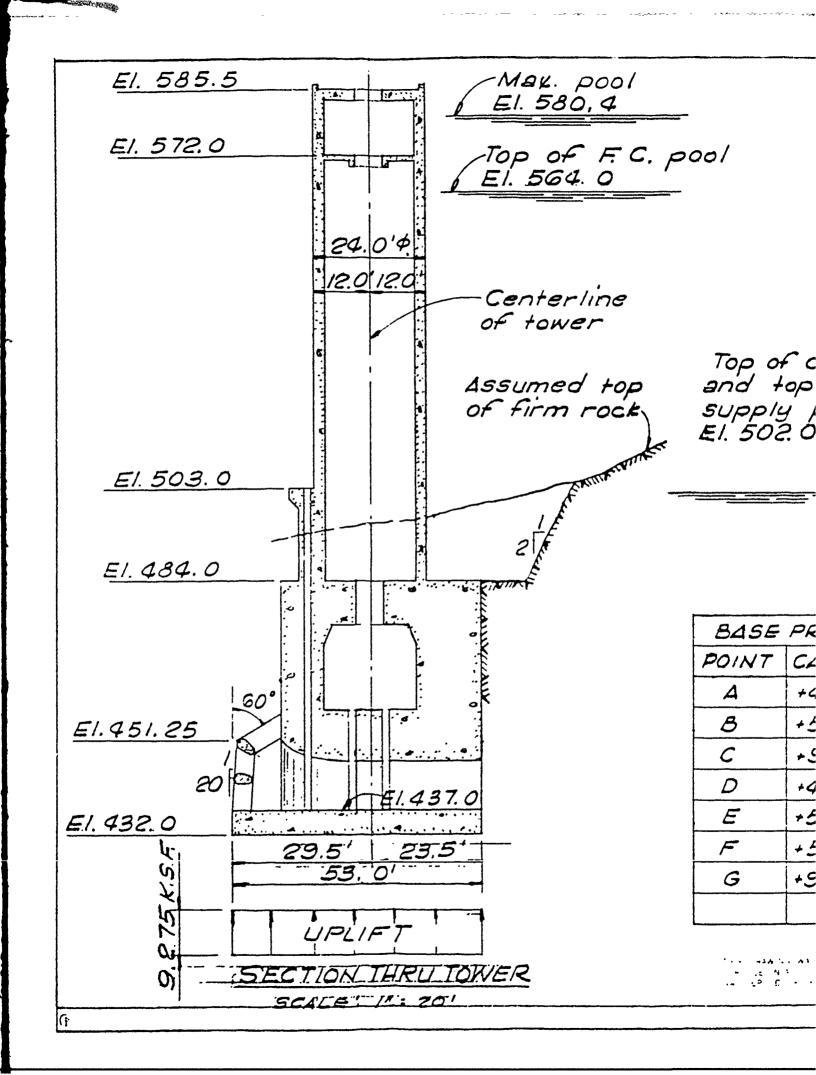












GILLHA MAK. POOL 1. 580,4 STAB DESIGN LOAD ASS op of F.C. pool AND BASE PRE i. 564. 0 CASE I (Construction co. Concrete only. CASE II Water to El. 580 service gate clos enterline acting over tota tower Top of conservation and top of water numed top DESIGN LO firm rock supply pool Concrete: Wc El. 502.0 Water! WI PI 53.0 27.721 BASE PRESSURES (KSF.) (\mathcal{B}) POINT CASE I CASE II A +4.97 + 1.84 4 85 +2.16 B +5.78 2.75 C + 3.71 +9.13 D +4.76 +2.11 25/ E + 2.51 +5.51. OF + 2.59 +5.45 2.00' +4.14 +9.14 10.0 17.721 PLAN OF SCALE: EMBANKMENT CRITE

GILLHAM GATE TOWER

DESIGN LOAD ASSUMPTIONS AND BASE PRESSURES

CASE I (Construction condition)

Concrete only.

CASE II Water to El. 580.4 with right service gate closed and left stop log closed and with 100 % uplift acting over total base area.

nservation of water

DESIGN LOADS

Concrete: Wc = 150 #/c.F. Water: W1 = 62.5 #/c.F. P1 = 62.5 #/s.F.

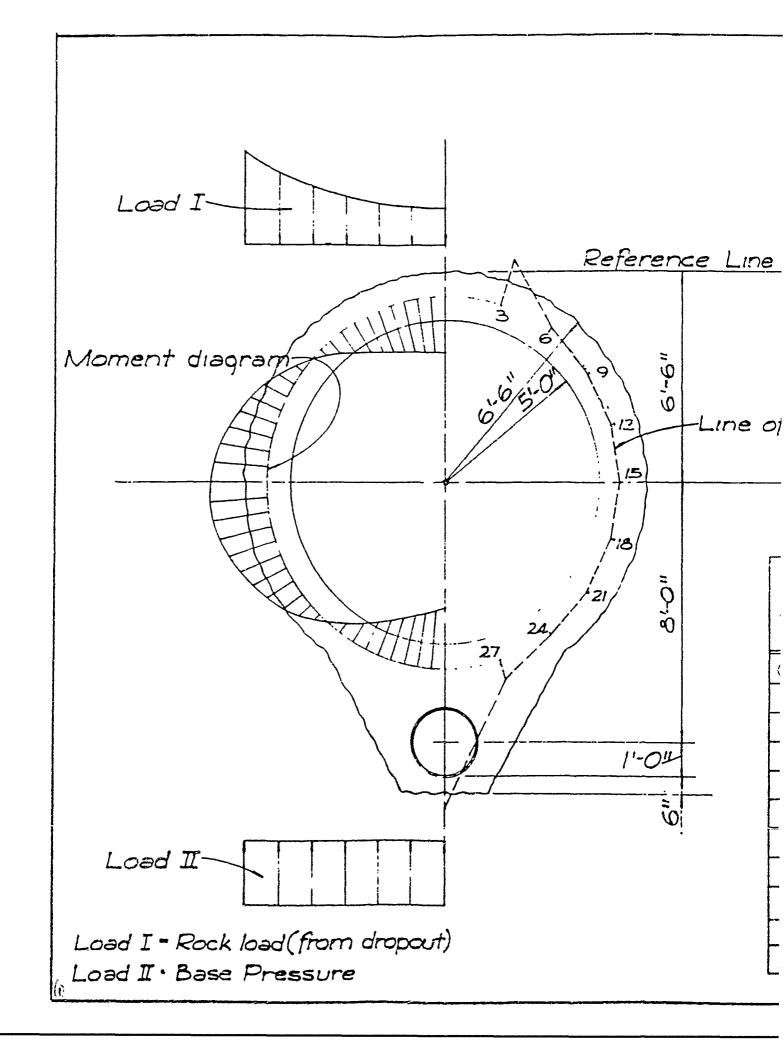
550	RES(KSF.
EI	CASEII
97	+ 1.84
78	+2.16
13	+ 3.71
76	+2.11
51_	+ 2.51
15	+ 2.59
4	+4.14

	53.		
, Vo	27.721	25.28'	
0		/	
	B	0	
18.85	(<u>A</u>)		8,70
99	<u>2.75</u>	EV: 16,69 Case I	7× 0,
	ion	Case I	4 20
1.15	0.	10	0 7
4	© 2.00'	EV=27,67 Case II	ZK B
	10.0 E 17.721	' Case II 25.28'	2 × 00 0
00.00	(F)	<u></u>	
	F77.4.1.	OF: F.405	

PLAN OF BASE

SCALE: 1" = 15'

(3)



Reference Line

1'-0"

Line of Reaction

GILLHAM.

Dimension Scale 30

Moment Scale 20

Line of Reaction Scale

40' 0

NOTES:

- I. Analysis presented represent severe moment loading that anywhere in the zone the section
- 2. Section was also checked again failure by low moment high three
- 3.All loads are computed on basis of Computer Program Number
- 4 Minimum steel of #6@12" will be section where steel is not other. This includes longitudinal steel

,						
Point	Moment kıp-ft.	Shear kıps	Thrust kıps	As Req'd	₿ars	As Supplied
Crown	11.81	0	2 <i>1</i> 6	0.606	#8e12	0.79
3	10.279	2.340	.534	.528	# <i>&</i> @12	C.79
6	4.992	3.538	2.522	.152	#5@12	0.44
9	-2.205	3.944	5.062	0	#6@12	0.44
12	-8.692	2.595	7.287	.180	#6@12	0.44
15	-12.037	.216	8.453	.309	#6012	0.44
18	-10.823	-2.374	8.005	.263	#6012	0,44
21	- 5.320	-4.095	6.003	.056	#6012	0.44
24	2.377	-4.202	3.319	0	#6@12	0.44
27	9.215	-2.628	1.081	.402	#6012	0.44
Invert	12.455	0	.216	0.639	#8@i2	0.79
	55	-				

THE MERINA WALL WARRENCE OF LARGE - -: 14 . V A RAN UV ANT

-GILLHAM DAM 10' + TUNNEL SECTION

Dimension Scale

Moment Scale

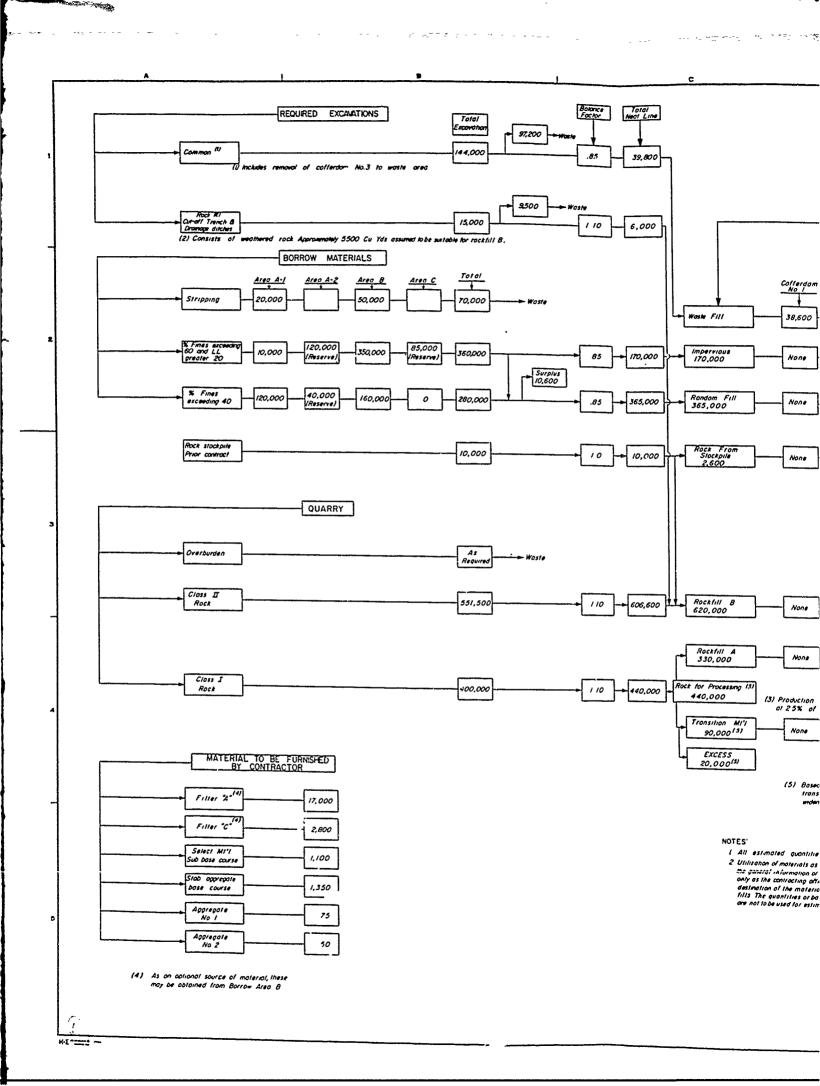
Line of Reaction Scale

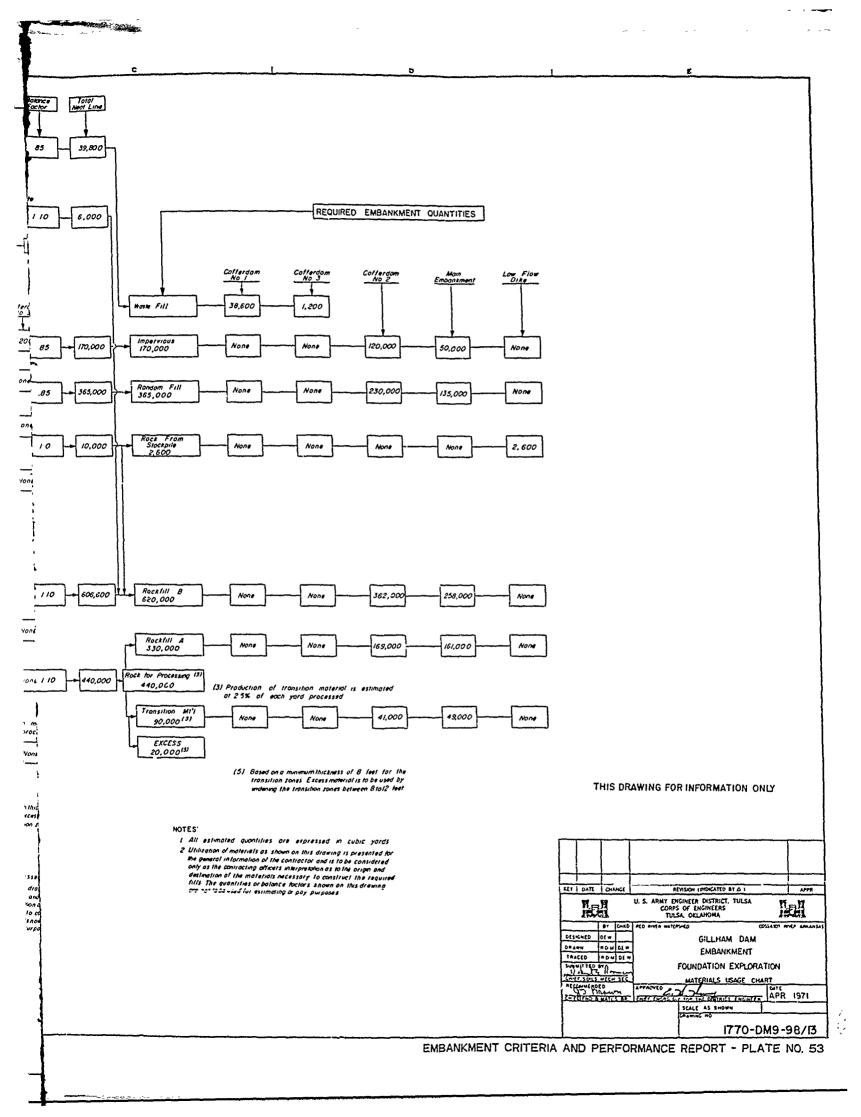
NOTES:

- I. Analysis presented represents the most severe moment loading that occurs anywhere in the zone the section is to be used.
- 2. Section was also checked against compression failure by low moment high thrust loads.
- 3. All loads are computed on basis of equations of Computer Program Number 13-G1-G515.
- 4 Minimum steel of #6@!2" will be used in section where steel is not otherwise shown. This includes longitudinal steel.

eaction

n#	Moment kıp-ft.	Shear kıps	Thrust kıps	As Req'd	₿ərs	As Supplied	fc 1800 psi . allow.	
wn	11.81	0	2/6	0.606	#8012	0.79	595	0
3	10.279	2.340	.534	.528	# <i>8</i> @12	0.79	518	16
0	4.992	3.538	2.522	.152	#6@12	0.44	288	25
7	-2.205	3.944	5.062)	#6@12	0.44	200	27
>	-8.692	2.595	7.287	.)	#6@12	0.44	549	2!
5	-12.037	.216	8.453	.309	#6012	0.44	804	2
;	-10.823	-2.374	8.005	.263	#6e12	0.44	7/2	-19
	- 5.320	-4.095	6.003	.056	#6012	0.44	299	- 33
4	2.377	-4.202	3.319	0	#6@12	0.44	125	-30
7	9.215	-2.628	1.081	.402	#6@12	0.44	573	-17
ert	12.455	0	.216	0.639	#8@i2	0.79	628	0
	<u>(2)</u>							130





END

FILMED

DATE:

DTIC